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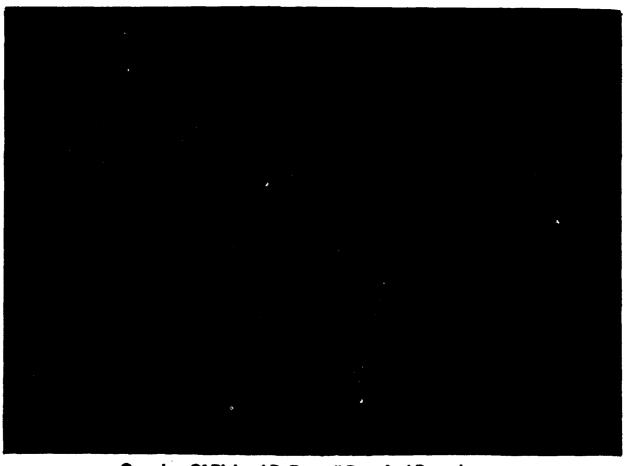
Richard B. Russell Dam and Lake

Savannah River, Georgia And South Carolina

DTIC ELECTE DEC 1 2 1989

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Final Foundation Report Volume 1 of 2



Overview Of Richard B. Russell Dam And Powerhouse



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DEPARTMENT OF THE ARMY

SAVANNAH DISTRICT, CORPS O ENGINEERS P.O. Box 889 SAVANNAH, GEORG!A 31402-0889

ATTENTION

CESAS-EN-B

30 November 1989

MEMORANDUM FOR Commander, Cameron Station, Alexandria, VA 22314, ATTN: DTIC/DA-2

SUBJECT: Richard B. Russell Dam and Lake, Savannah River, Georgia and South Carolina, Final Foundation Report

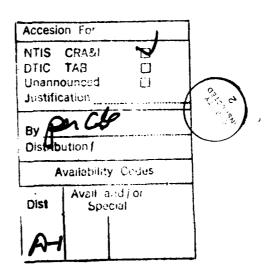
- 1. Enclosed are twelve (12) copies of referenced report, submitted in accordance with ER 1110-1-1801.
- 2. If you have any questions, please feel free to call the Project Manager, Mr. Bill Lynch, at 912-944-5701.

FOR THE COMMANDER:

Encls (12 cys)

oseph H. Rogers, jr., p.e.

Acting Chief, Engineering Division



SECURITY CLASSIFICATION OF THIS PAGE (When Date Entered)

REPORT DOCUMENTATION PAGE		READ INSTRUCTIONS BEFORE COMPLETING FORM
1. REPORT NUMBER	2. GOVT ACCESSION NO.	3. RECIPIENT'S CATALOG NUMBER
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16. DISTRIBUTION STATEMENT (of this Report)

Distribution of report was in accordance with ER 1110-1-1801

17. DISTRIBUTION STATEMENT (of the abstract entered in Block 20, If different from Report)

The abstract was written for Block 20 of this form which was included in all reports submitted to Defense Technical Information Center (DTIC) according to ER 1110-1-1801

18. SUPPLEMENTARY NOTES

The report was prepared from written naratives, drawings, and oral information given by project geologists Timothy A. Pope and Charles H. Combs.

19. KEY WORDS (Continue on reverse side if necessary and identity by block number)

Explorations and Investigations prior to construction

Regional Geology

Site Geology

Engineering Characteristics of soils and bedrock

Special design considerations

(See Reverse Side)

20. ABSTRACT (Continue on reverse side if necessary and identity by block number)

The report is a detailed description of the work involved in building of the Richard B. Russell Project. This includes the concrete dam, earth embankments, and the powerhouse. The report covers initial site exploration, geology of the site, special design considerations, excavation procedures, foundation exploration during construction, unusual conditions present in the foundation and foundation treatment. Grouting of the embankments and concrete dam is illustrated in numerous plates that show the various grout curtains present. The character of the foundation is shown in photos and foundation maps. The report

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19. (Continued)

Instrumentation

Possible Future Problems.

Excavation Procedures

Foundation Exploration During Construction

Unusual Conditions

Special Fault Zone Treatment; Francisch Variable

Foundation Anchors and Bolts: Employee Variable

Character of Foundation

Foundation Treatment

Grouting

20. (Continued)

Possible future problems are also addressed. Proceeding and after construction.

RICHARD B. RUSSELL PROJECT SAVANNAH RIVER, GEORGIA AND SOUTH CAROLINA FINAL FOUNDATION REPORT CONCRETE DAM, EMBANKMENTS AND POWERHOUSE

In Two Volumes

VOLUME I - TEXT AND PLATES

U.S. ARMY ENGINEER DISTRICT, SAVANNAH CORPS OF ENGINEERS SAVANNAH, GEORGIA

RICHARD B. RUSSELL PROJECT

SAVANNAH RIVER, GEORGIA AND SOUTH CAROLINA

FINAL FOUNDATION REPORT

CONCRETE DAM, EMBANKMENTS, AND POWERHOUSE

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RICHARD B. RUSSELL PROJECT

SAVANNAN RIVER, GEORGIA AND SOUTH CAROLINA

FINAL FOUNDATION REPORT

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RICHARD B. RUSSELL PROJECT

SAVANNAH RIVER, GEORGIA AND SOUTH CAROLINA

FINAL FOUNDATION REPORT

CONCRETE DAM, EMBANKMENTS, AND POWERHOUSE

PERTINENT DATA

LOCATION OF DAMSITE

Savannah River, Georgia and South Carolina, 275.1 miles above the mouth, 262.0 miles above Savannah, Georgia, 63.0 miles above Augusta, Georgia, and 37.4 miles above Clarks Hill Dam.

DRAINAGE AREA	Square Miles
Above mouth of Savannah River	10,579
Above Augusta, Georgia	7,245
Above Clarks Hill Dam	6,144
Above Richard B. Russell damsite	2,900
Above Hartwell Dam	2,088
Between Richard B. Russell and Hartwell Dams	812
ESTIMATED NATURAL STREAMFLOW AT DAMSITE	c.f.s.
Mean discharge for period of record (31 years)	5,078
Minimum discharge	
Instantaneous (11 May 1961)	300
Daily (24, 25 September 1961)	340
Monthly (May 1961)	969
Maximum discharge	
Instantaneous (25 August 1908)	114,000
Monthly (March 1952)	16,910
DESIGN FLOWS	c.f.s.
Standard project flood	
Peak reservoir inflow	518,000
Maximum estimated outflow	360,000
Spillway design flood	
Peak reservoir inflow	1,035,210
Maximum estimated outflow	801,500
LAKE ELEVATIONS	Feet, m.s.l.
Spillway design flood	490.0
Induced surcharge pool	485.0
Standard project flood	- 484.3
Flood control pool	480.0

LAKE ELEVATIONS (Cont'd)	Feet, m.s.l.
Maximum power pool Average power pool Minimum power pool	475.0 473.0 470.0
RESERVOIR AREAS	Acres
Standard project flood, elevation 484.3 Flood control pool, elevation 484.3 Maximum power pool, elevation 475 Minimum power pool, elevation 470 Limits of clearing, elevation 477 - elevation 465 Limits of clearing, elevation 465 - elevation 430 Limits of clearing, below elevation 430	31,770 29,340 26,653 24,117 5,880 11,890 7,540
RESERVOIR STORAGE VOLUMES	Acre-feet
Spillway design flood, elevation 490 Standard project flood, elevation 484.3 Flood control pool, elevation 480 Maximum power pool, elevation 475 Minimum power pool, elevation 470	1,488,155 1,305,000 1,166,166 1,026,200 899,400
ALLOCATED STORAGE VOLUMES	Acre-feet
Flood storage, elevation 475 - elevation 480 Power storage, usable, elevation 475 Surcharge storage, elevation 480 - elevation 490	140,000 126,800 322,000
TAILWATER ELEVATIONS	Feet, m.s.l.
Spillway design flood (801,500 cfs) Standard project flood (360,000 cfs) Maximum pool of record Four units operating at average head Maximum static condition	356.0 343.0 331.0 327.5 312.0
<u>DAM</u>	
Type: Concrete gravity and earth embankment Length, feet Concrete section Earth embankments Saddle dike Elevations, m.s.l. Top of dam	1,883.5 2,632.0 1,100.0 495.0
Streambed Spillway crest Top of tainter gates, closed	300.0 436.0 481.0

DAM (Cont'd)

Freeboard, feet	5
Height, maximum, feet	
Concrete section	205
Earth embankment	160
Top width, maximum, feet	
Concrete (nonoverflow)	20.7
Earth embankment	20.0
Side slopes, earth embankment	
Upstream	1V on 3H
Downstream	1V on 2.5H
Spillway	
Type: Concrete gravity ogee	
Gross length, feet	590
Clear opening, length, feet	500
Tainter gates, number, width, heights	10-50'x 45'
Type of bucket	flip
Radius of bucket, feet	50
Bucket lip, elevation m.s.l.	357.25
Intake section	35.72
Length, feet	
Conventional units	284.0
Pumped storage units	315.5
Total	599.5
Intake invert elevation m.s.l.	370.0
POWERPLANT EQUIPMENT	
Initial units, number	4
Ultimate units, number	8*
Net operating head, maximum, feet	162
Average operating head, feet	144
Minimum operating head, feet	134
Assumed head losses, feet	1
Turbine capacity per unit, hp, Average head	104,000
Maximum discharge per unit at minimum head, cfs	8,000
Unit spacing, feet	
Conventional units	71
Pump units	80
Transition spacing	75.5
Type of turbine	Francis
Elevation of distributor	300
Elevation bottom of draft tube	
Conventional units	257
Pump units	250
Draft tube exit elevation, conventional and pump units	265
Length of draft tube, feet	71
Generator capacity per unit, kW	75,000
Initial installation, total kW	300,000

POWER PLANT EQUIPMENT (Cont'd)

Ultimate installation, total kW Dependable capacity, initial installation, kW	600,000 300,000
Generator rating (95 percent pf), kVA	78.947
Generator speed, rpm	120
Generator voltage, kV	13.8
Transformer rating, kVA, Two 3-phase, each	182,000
Transformer voltages, kV	13.2 to 230
Average annual energy, million kWh	
At site	464.5
Prime	365.0

* Provisions have been made for the future installation of four 75,000 kW pump turbine units which will provide an additional 300,000 kW of capacity. An extensive system of rock bolting and rock anchorage was implemented for adding stability against uplift forces acting against the bases of units 5, 6, 7, and 8 (the future motor-generator bays). References to this special anchoring treatment are found throughout this text.

QUANTITIES

Concrete dam (nonoverflow and spillway)	
Excavation, common, cy	70,000
Excavation, rock, cy	105,000
Excavation, dredged, cy	209,000
Mass concrete, cy	562,000
Concrete, spillway piers, and training walls, cy	22,100
Concrete, terminal cone retaining wall	950
Reinforcing steel, lb.	3528,000
Tainter gates and embedded metal, lb.	2475,000
Stoplogs and guides, lb.	324,000
Sluice gates and embedded metal, lb.	290,000
Miscellaneous steel, lb.	18,000
Galvanized steel, lb.	26,000
Concrete dam (power intake section)	
Excavation, common, cy	19,000
Excavation, dredged, cy	212,000
Excavation, rock, cy	95,000
Mass concrete, cy	338,000
Concrete, intake headwork, cy	76,000
Reinforcing steel, lb.	4,446,000
Galvanized steel, lb.	45,000
Intake gates and guides, lb.	1,464,000
Trash racks and guides, lb.	1,483,000
Bulkheads and guides, lb.	440,000

- NOTES: (1) All elevations in this report refer to feet above mean sea level.
 - (2) Period of record August 1896 to present. Maximum discharge at Calhoun Falls, South Carolina, was computed for flood of 25

August 1908 as being 114,000 cfs. Maximum stage known since at least 1896, that of 25 August 1908, see U.S.G.S. Surface Water Records of Georgia 1969.

1.0 INTRODUCTION

1.1 Location and Description of Project: The Richard B. Russell Dam is located on the Savannah River near Calhoun Falls, S.C., 63 miles above Augusta, Georgia. It is 37.4 miles above Clarks Hill Dam and 29.9 miles below Hartwell Dam (see Figure 1). The Richard B. Russell Dam and Lake is a multipurpose project designed to provide hydropower, some flood control, recreation, and has a potential for water supply. The dam consists of a 195 foot high, 1,884 foot long concrete gravity structure (approximately 1.1 million cubic yards of concrete in 32 monolithic segments) flanked by two earth embankments (approximately 2.9 million cubic yards of zoned material.) The Georgia embankment is 2,180 feet long and the South Carolina embankment is 460 feet long. Within the powerhouse are four 75-megawatt conventional units and space for four 75-megawatt pump units. The project is designed as a peaking plant, with an installed capacity of 600 megawatts.

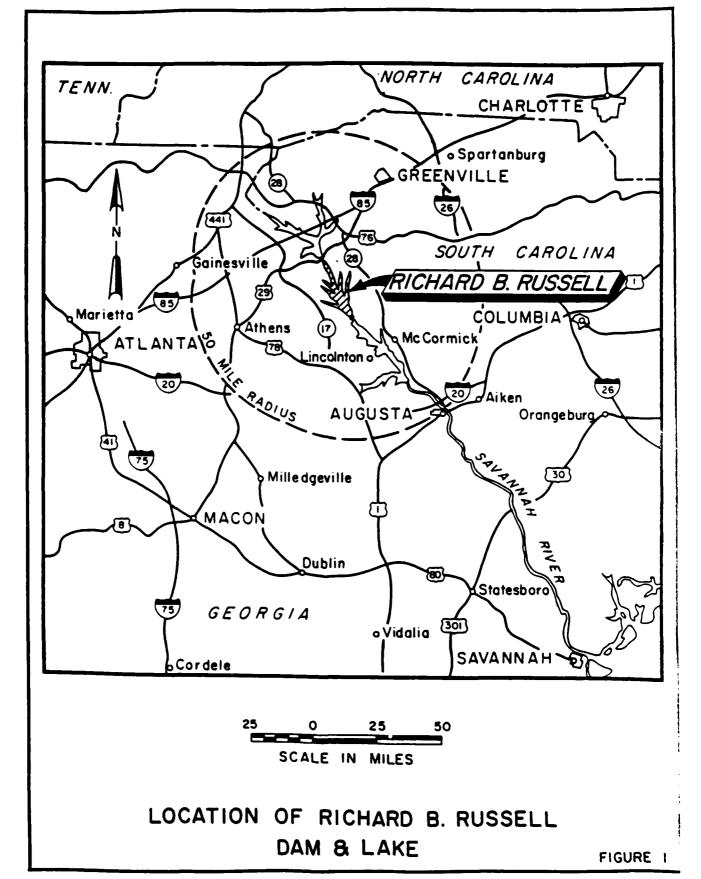
A general plan of the project is shown on Plate 3. The project plan provides for a maximum power pool at elevation 470 feet; spillway design flood pool is elevation 490. Tailwater elevations are: spillway design flood, 356; standard project flood, 343; and average operation (four units at average head), 327.5. The top elevation of both the concrete structure and the earth embankments is 495.

The concrete dam includes a spillway section containing ten tainter gates, an intake section, and a powerhouse containing the aforementioned 75 megawatt units.

The embankments are of rolled-fill construction. The Georgia (right) embankment extends 2,180 feet in a southwesterly direction. The South Carolina (left) embankment trends in an easterly direction for a distance of approximately 460 feet. Both of the embankments tie-in to the adjacent terrain (valley slopes) at elevation 495. Also located on the South Carolina (east) side of the dam is a 1,100 foot long saddle dike. Of similar construction to the earth embankments, the saddle dike was used to fill a swale or low-lying area of South Carolina State Road 269.

In addition to the generation and water retention structures, there were several highway and railroad bridges constructed for the project. Bridges which were founded on rock include the R-1 bridge, where the SCL railroad crosses the Savannah River, the Highway 72 bridge between Georgia and South Carolina, and the Georgia 368 - South Carolina 184 bridge near Iva, South Carolina.

1.2 Construction Authority: Basic project authorization is contained in the Flood Control Act of 1966 - Public Law 89-789, 89th Congress HR 18233, 7 November 1966. Authorization to include minimum provisions for the future installations of pumped storage facilities was granted by the Assistant Secretry of the Army in January 1977. Section 182a of the Water Resources Development Act of 1976 amended the authorization for the Richard B. Russell project contained in Section 203 of the 1966 Act by deleting the phrase which precluded the inclusion of pumped storage power as a part of this project and as a result design proceeded based on providing installation of pumped storage. House Document No. 97-35 dated March 19, 1981 was presented to the 1st Session of the 97th Congress by the Assistant



Secretary of the Army. This report was furnished to Congress to advise them of the Chief of Engineers' analysis and intention of proceeding with the construction of pumped storage hydropower facilities.

- 1.3 <u>Purposes of Report</u>: The purpose of this report is to provide a summary record of the actual geological conditions encountered during construction of the dam, powerhouse, and embankment foundations; the geotechnical methods and processes utilized to insure adequate foundations; and of any conditions which might lead to problems in the future.
- 1.4 Project History: Construction of the project began with the award of the contract for construction of the west access road to the A.J. Kellos Company in November of 1974. This road, which connects the Bobby Brown Park road to the damsite, was completed in 1975.

In December 1975, Southeastern Highway Contracting Company was awarded the contract for first stage diversion of the Savannah River. They were to excavate a channel into rock west of the intended concrete damsite, construct a bridge across the channel, build cofferdams upstream and downstream of the site, and divert the river through the new channel. During excavation, work was slowed by discovery of a prominent fault which crosses the west embankment area. This fault is described in detail in the Design Earthquake Report (1977) and will also be covered in Section 7.1-EE of this report. First stage diversion was accomplished in Spring 1977.

In August 1977, Lane Construction Corporation of Meriden, Connecticut was awarded the contract for excavation of the concrete dam and powerhouse area. In order to expedite the construction schedule, this phase of the work had been left as a separate contract prior to setting of powerhouse grade and completion of concrete dam design. Excavation began on the east abutment (Monoliths 27-32), then on the west abutment (Monoliths 1-7), then into the intake (Blocks 8-15) and spillway (Blocks 16-26) areas as work was being completed on the east abutment. Excavation in the powerhouse area was being completed as the concrete dam contractor was beginning his trestle construction.

Final elevations in several areas of the excavation were revised as construction progressed. These areas included Monoliths 2 and 3 (where a shallow pocket of decomposed rock was removed with a backhoe), Blocks 16-21, (underlain by a relief joint filled with decomposed rock materials 3 to 5 feet below design grade), and Blocks 26-30, where clay-filled joints and a wide shear zone affected rock quality. Monoliths 27 and 28 required the most extreme lowering, being excavated 15 to 38 feet below original design grade. The work on these two blocks was completed under modification after excavation of the powerhouse area, in the summer of 1979. These changes are depicted on Plates 13 and 16.

The concrete dam contract was awarded to Dravo-Groves (a joint venture) in May of 1978. Preliminary site work began soon after the notice to proceed was issued on 16 June 1978. The batch plant (four, 4.5 cubic yard mixers) was built just west of the erection bay. Aggregate handling facilities were built (some blasting was necessary) in the present switchyard area, and a trestle system was begun in the Geogia abutment area with footers 85 and 121 feet downstream of the dam axis.

Suitable material for aggregate (for the most part identical to the foundation rock types) was located north of the S.C. 269 saddledike location, less than a mile northeast of the dam. The quarry was developed and tested in the fall and winter of 1978 and aggregate production began in January 1979. The original crusher setup, which depended on a roller crusher, was inefficient and had to be replaced in the middle of construction. This change is described in the Concrete Report.

By March 1979, the trestle system had been extended into the intake blocks, a few monolith foundations were being prepared, and the batch plant was ready to produce mass concrete. Where foundation relief varied by more than 5 or 6 feet, "plug pours" were usually made to level-up the placement area. The first mass concrete in the dam structure area was placed in the Monolith 2 plug pour on 19 March 1979. The last placement on a rock surface was completed in Block 32 on 15 February 1981.

Drilling and grouting operations for the concrete dam foundation began in December 1980, and continued through October 1981. A single line, 3 zone curtain, 120 feet deep having holes on 5-foot centers was installed. Some split-spaced holes between those centered every 5 feet were also grouted. Problems which complicated or delayed this work were: leaks through the concrete onto the gallery floor, leaks through the concrete dam into the earth embankment areas, and interception of the inverted plumb bob in Monolith 26 with a grout hole. The contractor experimented with a down-the-hole percussion type hammer in drilling the foundation drain holes, but when this proved inefficient, he completed the remainder with rotary equipment. The rock (especially the metadacite) proved to be very hard to drill, making plug bits less economical than coring bits. The subcontractor (Boyles Brothers) experimented extensively with different types of bits, finally settling on impregnated coring bits. Refer to Plate 50 for details of drain and grout hole drilling.

Instrumentation was installed throughout the job. Difficulty was experienced in getting drills positioned for drilling inverted plumb bob, uplift cell, and deflectometer holes due to the uneven foundations. These holes (with the exception of the deflectometer holes) were drilled by subcontractors, making scheduling a problem for the contractor. If they were drilled before foundation work, rock removal could cut as much as 5 feet from the original drilled depth. If the foundation was prepared first, drilling could delay concrete placement. Finally, the contractor decided to do some rock removal in the general area of the holes prior to drilling them. Instrumentation that was installed included 33 uplift cells in Block 7, 10 cells in Block 10, 33 in Block 23, 10 in Block 26, 19 in Block 27, 23 in Block 28, and 6 in Block 29; plumb bobs and inverted plumb bobs in Monoliths 16 and 26, and deflectometers (1 each) in Monoliths 27 and 28. The deflectometers were installed in the foundation with a stub-up to gallery level. Then concrete was placed to full height with a 6-inch blockout around the deflectometer location. After this, the deflectometer was assembled and installed from the top of the dam, connected to the stubup in an opening left in the gallery for this purpose, and the block-out grouted up. All work in the concrete dam was considered essentially complete in June 1982.

- 1.5 Location of Structure: The concrete dam is positioned perpindicularly across the Savannah River channel. Blocks 1-9 are set into the old Georgia riverbank, and Blocks 27-32 are set into a bluff which was the old South Carolina riverbank. Intake Blocks 8-16 and spillway Blocks 16-26 span the old riverbed. The powerhouse is located immediately downstream (south) of the intake area. The concrete structures are situated between the Georgia and South Carolina earth embankments, both of which are founded well above the old riverbed elevation. The earth embankments trend generally in a northeasterly-southwesterly direction, whereas the concrete structures follow a nearly east-west alignment. The deepest part of the river channel was in the vicinity of Blocks 20-22. These spatial relationships are depicted on Plate 3.
- 1.6 Contractors, Supervision, and Quality Control Organizations: The A.J. Kellos Company built the West Access Road. The Project Manager/Job Superintendent was Mr. McCord.

The contractor on the first stage diversion contract was Southeastern Highway Contracting Company of Gainesville, Georgia. James E. Trammel served as Project Manager/Job Superintendent.

Lane Corporation of Meriden, Connecticut was the contractor for the concrete dam and powerhouse excavation contract. The Project Manager/Job Superintendent was Mr. Darrel Emig. The Quality Control Organization consisted of the Quality Control Chief, Mr. Robert Halpin, and members of the supervisory staff, primarily Mr. H.W. Anderson and Neil Armstrong.

The concrete dam, as well as the earth embankments, were built by Dravo-Groves, (a joint venture of the Dravo Corporation of Pittsburg, Pennsylvania and S.J. Groves and Sons of Minneapolis, Minnesota). Excavation of the cut-off trench beneath the earthen portions of the dam was also handled under this arrangement. The Dravo Corporation was responsible for supervision and operation of the construction forces. The project manager was Mr. Joseph J. Smith. The Job Superintendent was Charlie Ramsey. The Quality Control Organization was composed of the Quality Control Chief, Doug Sickle, and members of the contractor's and subcontractor's supervisory staff. Foundation and instrumentation work were monitored by Mr. Sickle, J.C. Wallis, Jim Adkins, and Roy Stone of Dravo. Grouting and all drilling operations were monitored by Steve Hall of Boyles Brothers (subcontractors). Installation of instrumentation in the dam was monitored by Andy Anderson of Martyn Brothers (subcontractor).

The joint venture of Replublic-Crowder was awarded the powerhouse construction contract in August of 1981. Some modification to the previously excavated foundation surface was also handled by this concern.

1.7 Key Resident and Design Staff:

Key Resident Staff:	Resident/Area Engineer Asst. Resident Engineer Asst. Area Engineer	-	1976-1982 1977-1979 1979-1982
	Project Geologists	Robert H. Stephens James E. Hastings	1976-1977 1977-1978

	Project Geologist	Timothy A. Pope	1979-1982
	Staff Geologists	Timothy A. Pope Mary Lou Fagan Charles H. Combs	1978-1979 1979-1981 1981-1982
	Technicians	Jack Ford Thomas B. Keane James R. Zanders	1978 1979-1980 1980-1982
Key Design Staff:	Geologists	Robert S. Stansfield Charles M. Deaver Earl F. Titcomb, Jr. Ron M. Rhodes	1974-1979
	Engineers	Ben C. Foreman Tom Durrence John Hager	1974-1984 1982 1984

2.0-CD CONCRETE DAM FOUNDATION EXPLORATIONS (PRIOR TO CONSTRUCTION)

A number of subsurface exploratory programs were carried out prior to construction. Among these were:

2.1-CD <u>Site Selection</u>: Borings A-1 through A-18: These borings were located along a baseline designated by the letter "A" which at one time was being considered as a potential axis for the concrete dam. Drilling of these borings was carried out between February and June of 1968.

Borings B-2, B-4, B-5, B-6, B-7, B-8, B-10, B-12, B-13, and B-15: These borings were located along a baseline designated by the letter "B" which also was under consideration as a possible dam axis. Drilling of these "B" prefixed borings occurred between March and May of 1968.

Borings C-1 through C-24: These borings were taken along a baseline designated by the letter "C". Drilling was conducted between March and June of 1968. Baseline "C", with slight modification, ultimately became used as the concrete dam axis.

Borings 1, 2, 3, and 4 were also drilled in conjunction with the site selection process and their locations were relative to baseline "C". These were completed during October and November 1964.

Borings PF-1, PF-2, PF-3, PF-3A, PF-4, PF-6, PF-7, and PF-8 were drilled relative to baseline "A" in November and December 1961.

Borings PQ-1, PQ-2, PQ-3, and PQ-4 were drilled between November and December of 1961 in an investigation of potential quarry sites for rock aggregate.

Geophysical testing such as cross-hole sonic velocity measurement and Schmidt Hammer rebound testing was also performed, the results of which are presented under Section 3.5: "Engineering Characteristics of the Bedrock Materials".

Plate 4 shows the linear arrangements of the potential baselines once considered for the concrete dam.

2.2-CD <u>General Design</u>: Following selection of baseline "C" as the approximate location of the dam axis, a number of additional borings were made, the locations of which were relative to this baseline. Borings C-25 through C-96, drilled between September of 1968 and August of 1969 are among these.

Other investigations, which concern the embankments, saddle dike, service-road fill, powerhouse, and potential borrow areas will be highlighted later in this text.

2.0-EE EARTH EMBANKMENT EXPLORATIONS (PRIOR TO CONSTRUCTION):

2.1-EE Foundation Explorations: The foundations for the Georgia and South Carolina embankments were explored by 90 and 51 borings, respectively. Also, on the South Carolina side, 11 additional borings were drilled in the

area of the service road embankment and 7 borings drilled along the axis of the saddle dike. Locations of all borings made for the Georgia and South Carolina embankment foundations are among those shown on Plates 58 and 59, respectively. The majority of the borings were continuously split-spooned through the overburden to refusal and then core drilled to obtain either 2.5-inch or 4-inch core samples. Selected 3 and 5-inch Shelby tube undisturbed samples were obtained from the overburden soil for laboratory testing. Selected 4-inch core samples were obtained of the saprolite (intensely weathered rock) and wrapped in cheesecloth and coated with wax to preserve them for laboratory consolidation and triaxial testing. Of the core borings, 43 were hydraulically pressure tested. In addition, the deep diversion channel excavation crossing the proposed alignment of the Georgia embankment served as an excellent inspection trench by exposing to view some of the rock types, geologic structures, and rock weathering that would be encountered in the cutoff trench.

- 2.2-EE Materials Explorations: A total of 182 borings and 15 backhoe test pits were made in the search for impervious borrow. Borings were made by hand auger, standard split-spoon, "square" or Vicksburg type auger, and spiral auger. "Four-by-five" double-tube core borings were made in attempts to locate intensely weathered rock. Soil-type borrow material was sampled with the hand auger, the split-spoon sampler, and "square" auger. The intensely weathered rock (saprolite) was sampled with spiral augers to determine the probable depths to which ripping could be accomplished. From the explorations conducted for the excavated diversion channel it was found that the spiral auger sampling provided a fairly good indication as to the depths of rippable material. Jar samples were taken in the borings for determination of in-situ moisture content, and bulk samples were obtained for laboratory testing. The river sand was investigated by standard split-spoon borings. Furthermore, an extensive test fill program was conducted for the impervious soil borrow and the sand in the dredged sand stockpile.
- 2.3-EE Groundwater Investigations: Groundwater levels recorded in exploratory borings showed that the groundwater at the site were effluent to the Savannah River. On the Georgia side, the water table was found to be above, or coincident with, the top of rock. On the South Carolina side, where the topography is steeper, the water table was generally found to be coincident with or below the top of rock.
- 2.4-EE Laboratory Investigations: Jar and bulk samples from the proposed impervious borrow area, bulk samples of the intensely weathered rock and dredged sand, and jar and undisturbed samples from the embankment foundations were tested in the South Atlantic Division Laboratory.

 Mechanical and hydrometer analysis, moisture content, and Atterberg limits were performed on the jar samples. In addition to the above tests, the following tests were performed on bulk samples of impervious and intensely weathered rock: standard compaction, consolidation, permeability, direct shear, and triaxial compression. Additionally, x-ray diffraction, pinhole erosion and Soil Conservation Service laboratory dispersion tests were performed on the impervious borrow material. Tests on the dredged sand included mechanical analysis, specific gravity, standard and modified compaction tests, maximum-minimum density tests, permeability, direct shear, and triaxial compression.

- 2.5-EE Dynamic Analysis: The proposed embankments were subjected to a dynamic stability analysis performed by the Waterways Experiment Station utilizing results from exploratory drilling, sampling, and field geophysical studies. Undisturbed samples were obtained from the foundation residual soil and saprolite and from test fills constructed of impervious material and dredged sand for appropriate dynamic testing. Bulk samples of the impervious, intensely weathered rock, and dredged sand were also obtained for appropriate dynamic testing. Dynamic testing of the materials was conducted at the South Atlantic Division Laboratory.
- 2.6-EE Clay Seams Study: The excavation for the concrete dam in the South Carolina abutment exposed mud and clay-filled seams in the rock. The seams first appeared as the excavation was approaching initial design grade, and it appeared that seams might be found below grade. Through the use of the exploratory borings and field inspections of the excavation as the monoliths reached design grade, the problem of the clay seams was carefully evaluated and addressed to ensure adequate and safe foundations for the concrete structure. Based on extensive field and laboratory investigations, it was concluded that the clay seams were not very continuous and did not pose a threat to the integrity and safety of the earth embankment. Because of the presence of the clay seams, the possibility of a portion of the abutment immediately upstream of the embankment sliding into the reservoir, and creating a wave which might overtop the dam was evaluated. But considering the facts regarding the clay seams, and since the abutment will be inundated by the reservoir, such an occurrence was considered a near impossibility. However, because there appeared to be a considerable number of seams in certain areas of the steeply dipping foundation for the downstream terminal cone, stability analyses were performed for the terminal cone assuming clay seams were completely continuous through the foundation. The analyses reinforced the conclusion that the clay seams pose no threat to the embankment.
- 2.7-EE <u>Filter Tests</u>: Laboratory filter tests were conducted by WES to establish the <u>adequacy</u> of the proposed sand as a filter, both in regard to its internal stability and its effectiveness in protecting the impervious core material. For conservatism, these tests utilized some of the finest core material and some of the coarsest sand. The tests indicated that stockpiled sand would adequately protect and filter the impervious core material.

2.0-PH POWERHOUSE FOUNDATION EXPLORATIONS (PRIOR TO CONSTRUCTION)

2.1-PH Previous Investigations: Geologic investigations made prior to construction are described fully in Design Memorandum 8 - Geology, General Design Memorandum 1, and Design Memorandum 12 - Materials. Geologic conditions of the project are described in detail in Section 3.0 of this report, and summaries of the geologic/geophysical testing programs may be found in Section 3.5 and Appendix A of this report. The 1977 COE report entitled "Geological Seismological Evaluation of Earthquake Hazards at the Richard B. Russell Project" covers in detail the regional and local seismicity of the project area and an evaluation thereof.

3.0 GEOLOGY

3.1 Regional Geology: The Richard B. Russell Project is in the Piedmont Physiographic Province. The area is broadly rolling with generally long, rounded stream divides. In the Southeastern Piedmont, the country rocks are commonly interbedded, metamorphosed, volcanic, and sedimentary units intruded by dikes and sills. After deposition, these rock layers were folded, faulted, and metamorphosed in several tectonic and metamorphic events. Plutons and dikes intruded the rock mass after metamorphism.

The lower Piedmont in Georgia and South Carolina can best be characterized as a series of NE-SW trending bands of alternating high and low grade metamorphic rocks (see Figure 2). Folding in these rocks is complex and variable with major fold axes trending NE-SW. The lower grade bands are usually metamorphosed to greenschist grade, with well-preserved depositional features common in sedimentary and some pyroclastic rocks. The low-grade bands are called the Carolina Slate Belt and Kings Mountain Belt. The Little River Series of Georgia (type locality is the Little River area on Clarks Hill Lake) is a continuation of the Carolina Slate Belt. The higher-graded bands (usually amphibolite grade) are referred to as Charlotte Belt rocks. In these rocks, preserved depositional features are uncommon, with metamorphic foliation the dominant fabric in pelitic rocks.

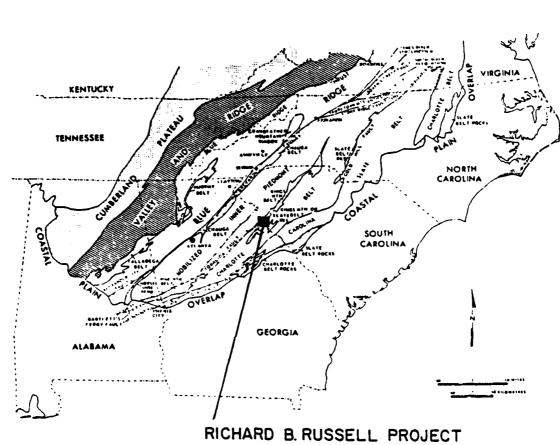
Faulting in the region was summarized in the Design Earthquake Report (COE, 1977) and Table 1 lists information on the faults of major interest. Some of these are shown in Figure 3 and none of them is considered to be active. Due to the presence of faults and location of the project within 170 miles of the Charleston earthquake of 1886 and occurrences of minor tremors in the vicinity, the dam was designed to withstand an earthquake of 5.5 Richter force with a shallow hypocenter at the damsite. The seismic-defensive design resulted in more complex zonation in the concrete edifice, a conservative approach to foundation preparation, and increased instrumentation.

3.2 Site Geology:

a. <u>Physiography</u>: The Savannah River Basin is long and relatively narrow. The Richard B. Russell Lake extends from the dam approximately 29 miles upstream, almost to Hartwell Dam. Maximum width at any point is approximately 3 miles.

The concrete dam is located in a bend of the river in the head waters of Clarks Hill Lake. The river width at the damsite is approximately 1000 feet at elevation 330 MSL. The Georgia riverbank is somewhat lower than that in South Carolina, necessitating a 2180-foot long west embankment. The South Carolina abutment of the concrete dam is located on a high bluff with only a 460-foot embankment needed to impound the lake. Maximum relief at the site (above the old riverbed) is about 250 feet. The lake is approximately 175 feet deep at its deepest point.

b. <u>Description of Overburden</u>: The Richard B. Russell area typically has a normally developed soil horizon, but in some areas the A horizon is absent and the B horizon attenuated. Residual soils vary in color from yellow or tan to dark red. The most common residual soil type is a brick

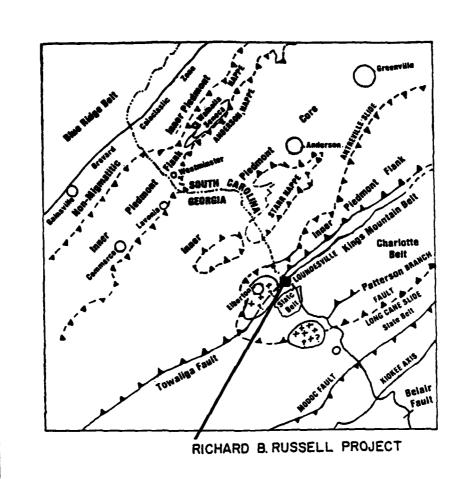


TECTONIC ZONES IN THE PIEDMONT

ADAPTED FROM HATCHER (1972)

FIGURE 2

	HINIMUM MAPPED LENGTH		PROBABLE	: AGE		
FAULT	(111)	FAULT TYPE	(MAX)	(nin)	STRIKE	APPROX LOCATION FROM DAMSITE
Modoc	79	1. Thrust 2. Strike-slip	Paleozoic	Late Devonian to Permian	NE-SW	30 miles SE
Lowndesville	160	Questionable Right Lateral Strike-Slip	Paleozoic	Late Devonian	NE-SW	15 miles NW
Brevard	450	 Right Lateral Strike-Slip Dominantly Dip-Slip 	Paleozoic	Late Devoni an	NE-SW	75 miles NW
Belair Fault	13	Reverse	50 H.Y.B.P.	Quaternary	NE-SW	50 miles SE-West of Augusta, GA
Patterson Branch	4.5	Normal or Thrust		Mesozoic	NE-SW	McCormick Co, SC approx 12 mi SE
Diversion Channel Faul	t 4/10	Thrust	355 H.Y.B.P	270 M.Y.B.P	N-S	Under west embankment



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TECTONIC MAP OF THE PROJECT AREA

ADAPTED FROM C.O.E. DESIGN EARTHQUAKE REPORT (1977)

FIGURE 3

red, readily erodible silt with a variable sand component. Most samples taken during exploration were classified as SM, SC, ML, or CL. Silica hardpen (formed from a weathering product of feldspers) can be seen in the area, but is usually poorly developed compared to that occurring in the Lincolnton, Georgia area.

Alluvium is confined primarily to the riverbed and its margin. River sand is normally a fairly clean sand composed mostly of subrounded quartz particles with some mica flakes and feldspar fragments. Sand and rounded cobbles in near vertical, open joints caused problems during drilling and blasting operations in the tailrace area.

Hard rock outcrops are rare in the region and, when found, are usually located along streams. Rock is commonly covered by 25 to 100 feet of overburden. Most of this thickness is saprolite occurring in the C and D soil horizons. Due to the rapidity of weathering in the southeast, this material shows relict structure which can often be used to identify the parent rock.

c. Bedrock Stratigraphy: The Richard B. Russell Dam is located in an isolated outlier of low-grade metamorphic rocks within the Charlotte Belt. The rocks have been referred to variously as Carolina Slate Belt, Kings Mountain Belt, and Little River Series lithologies. Rocks mapped in the concrete dam foundation are listed in the following paragraphs in order of probable age of emplacement. These rocks were identified in hand sample with the help of previous microscopic patrographic reports done for design memoranda and reports from a master's thesis done at the University of South Carolina by Weisenfluh (1977). Figure 4 provides further information about stratigraphic relationships of the area.

C

Xenoliths - A few xenoliths were mapped within the metadacite, mostly on the South Carolina side of the river. They are mafic-intermediate, fine-grained, clive colored rocks with some dark green, corroded phenocrysts of indeterminate mineralogy. They may represent the original rock mass intruded by the metadacite. The rock type is probably meta-andesite.

Metadacite - This is the major rock type found in the dam foundation area. It is a felsic (light-colored with quartz and feldspar as the major constituents) rock, generally of medium to coarse grained, with bluish quartz and cream-colored feldspar making up 85-90% of most samples. Accessory minerals are chlorite and/or biotite with small amounts of epidote, calcite, magnetite, pyrrhotite, and other opaques. It is indistinguishable in hand sample from the Lincolnton metadacite, which occurs about 20 miles south of the damsite. This rock, considered to be the oldest foundation rock, is referred to as quartz feldspar gneiss on maps and core logs.

Quartz Feldsper Porphyry - A fine-grained, dark gray rock with larger (up to 1.6 cm), subrounded phenocrysts of light green, saussuritized plagicclase feldsper and slightly opalescent, bluish quartz occurs in elongate bodies within the metadacite country rock sometimes bordered by metadiabase diles. On maps, this rock is called quartz feldsper porphyry. Weisenfluh (1977) refers to this rock as a quartz heratophyre (Ne+metasometized, submerine, volcanic intrusion). He describes the mineralogy

	TERTIANY CRETACEOUS JURASSIC TRIANSIC T	126 225 246	ROCK SE GEORGIA TUSCALOGRA FORMATION GRABEN BASIN FILL RHYOLITE PLOWS AND DRICES WACKE CONGLOMERATE SEQUENCE	OUENCES SOUTH CAROLINA TUSCALOGEA PORMATION GRAMEN BASIN FILL	PLINTY CRUSHED ROCK DIABASE DRIES GOSHSH GRANITE	METAMORPHIC EVENTS METAMORPHIC EVENT	OUTSIDE SO MILE RADRIES OF DAME
O COCOCOCO	CRETACEOUS JURASSIC TRIASSIC TRIASSIC PERMISPILVANIAN INTERNSPILVANIAN INTERNSPIPANI	65 135 195 225 230 345	GRORGIA TUBCALOGRA PORMATION GRABEN BASIN PILL RHYOLITE PLOWS AND DIKES	TUBCALOORA PORMATION	PLINTY CRUSHED ROCK DIABASE DIGES GOSHEN GRANITE	METAMORPHIC EVENT	RADRUS OF
O COCOCOCO	CRETACEOUS JURASSIC TRIASSIC TRIASSIC PERMISPILVANIAN INTERNSPILVANIAN INTERNSPIPANI	136 166 225 200 345	GRADEN BASIN FILL RHYOLITE PLOWS AND DUCES		DIABASE DIKES	METAMORPHIC	
	JURABBIC TRIABBIC PRIMILAN PRIMILAN INCOMPLETAN INCOMPLE	105 225 200 345	GRADEN BASIN FILL RHYOLITE PLOWS AND DUCES		DIABASE DIKES	METAMORPHIC	
	TRIABEIC PERMAN PERMAN INCOMPRAN IN	225 280 386	RHYOLITE PLOWS AND ORCES	GRABIN BASIN PLL	GAANITE	METAMORPHIC	
	PRINCIPLYANEAN INTERCEPPIAN ELYT INTERCEPTAN ATRICE ATRICE	200 346	RHYOLITE PLOWS AND ORCES		GAANITE	METAMORPHIC	
	BARLY REDUCE LATE	346			, 	METAMORPHIC EVENT	
	EARY MEDGE	386			, 	METAMORPHIC EVENT	•
	EARY MEDGE	365	WACKE CONGLOMERATE		, 	METAMORPHIC EVENT	T
	EARLY	365	WACKE CONGLOMERATE		, 	METAMORPHIC EVENT	
ONC	ESELV	386	WACKE CONGLOMERATE		, 	METAMORPHIC EVENT	
900		386	WACKE CONGLOMERATE		, 		
36 0			WACKE CONGLOMERATE	i	GARGEC		1 1
9	<u>•</u>		SEQUENCE		GABBRO, DIORITE SYRRITE COMMISSIES		
	Ħ	437			SYBUTE COMPLEXES ELBERT GRANITE		
PALEOZOK	1					METAMORPHIC AND	
	ORDOWCIAN					DEFORMATIONAL	
					 		MDGE -
	ELARLY		SEQUENCE SEQUENCE	DELAIR DELT	MEDIUM GRAMED ADAMELLITE		ma =
		500					- VALLEY
	. ~		5				
	3700 B		PROGLASTIC				
	" —		5		j		
	3		LINCOLNTON				
3	- -	3/4	(MACOGCHER)	COMMORBVILLE	SERPENTINE & ULTRABASIC		
	•		DADEVILLE	KINGS MQUNTAM			
	9		UCHEE BELT	CHARLOTTE BELT	ADMINISTR	METAMORPHIC EVENT	
	PRECAMENANT	CAMPELLAS CAMPELLAS EARLY MODILE LATE EARLY	CAMMENLAN CAMMENLAN LATE EARLY MOGLE LATE EARLY	See SECURITARY (SECURITARY (SECURITARY (SECURITARY (SECURITARY (SECURITARY (SECURITARY SECURITARY S	STOREMENTARY SECURITY SELAM SELT STOREMENTARY SELAM SELAM SELT STOREMENTARY SELAM SELAM SELT STOREMENTARY SELAM SELT STOREMENTARY SELAM SELT STOREMENTARY SELAM SELT STOREMENTARY SELAM SELAM SELT STOREMENTARY SELAM SELAM SELT STOREMENTARY SELAM S	STORMETON OF THE SECURITY SECU	SECURITY SEC

of the microcrystalline ground mass as quartzo-feldspathic, with sericite mica and accessory epidote, sphene, biotite, and chlorite. Field relationships indicate that this rock intrudes the metadacite and is cut by metadiabase diess.

Mylonite - A very fine-grained, dark gray rock which may be primarily quartzo-feldspathic in composition (minor weathering produces a white rind) was found in the spillway foundation and in Blocks 25-27. The dark color is probably due to the fineness of the translucent grains. This rock occurs in near vertical, tabular bodies cut by dark green metadiabase dikes. Foliation is evident in some parts. This rock was interpreted in the field, without microscopic study, as ancient mylonite. It may be older or younger than the quartz feldspar porphyry.

Metadiabase (or Metabasalt) - This is a mafic (darker in color with an abundance of iron- and magnesium-rich minerals) rock, generally of fine to medium grain size. It is a metamorphosed diabase or basalt, and is referred to on maps and logs as epidote hornblende gneiss. In some cases, this rock may not contain enough epidote to warrant the use of the mineral in the rock name, but this is not discernable in hand sample. Hornblende and feldspar are the major minerals in the rock with minor epidote, chlorite, quartz, pyrrhotite, magnetite, and pyrite.

Variations in the epidote hornblende gneiss should be mentioned. Petrographic reports on this rock show variable mineralogy, and crosscutting relationships indicate that there are several generations of dike intrusion. The material varies in color from dark green to black. Some dikes, especially thin ones and those parts of thicker dikes which have been sheared, show pronounced foliation, probably in ting a higher chlorite content. The term "chloritic epidote hornblende gneiss" has been used on the geologic maps in this report to indicate this type of rock. The NNW-SSE trending dike in Monoliths 5 and 6 is a green-black, fine-grained material with quartzo-feldspathic inclusions which have been partially recrystallized, possibly from the intense heat of the intruding mafic melt. Some of the inclusions had alternating light and dark gray beds or bands. Another variation in the type of dike mapped is epidote hornblende gneiss with phenocrysts of feldspar, quartz, or hornblende. The quartz phenocrysts in these dikes are similar in appearance to those in the metadacite and quartz porphyry, possibly indicating hybridization of the intruding magma. These dikes cut and were cut by metadiabase dikes (with no phenocrysts) which were indistinguishable from each other in hand sample.

d. Bedrock Structure, Jointing, Discontinuities: Structural features in the dam foundation include several types of faults. The most common is a simple, clean break along which there is a small displacement. Due to the limited displacement on these faults, it is difficult to tell if they are strike or dip-slip faults, but some indicate both types of movement. These breaks are near vertical, commonly trending NM-SE, and are of limited extent. Similar, but longer faults occur in Blocks 2 through 7. These faults (partially lined with epidote) trend NE-SW, have variable, near vertical dips, and show slickensides with well-developed striations parallel to the dip direction. Low angle faults, healed by zuned quartz and chlorite veins, also occur in the foundation. These are apparently reverse faults of

35 degrees or less dip. They offset some metadiabase dikes and are themselves offset by (visually) identical dikes.

Planes and zones of shear were also mapped. These zones vary from less than an inch to more than a foot thick. They commonly cut along dikes, but are also found sheared through the country rock for short distances. In addition to crushed and altered rock, these zones contain chlorite, calcite, a pink zeolite, gouge, and other materials of hydrothermal origin. Sense of movement along these shear zones is commonly indeterminate, with evidence of multiple movement direction seen in several zones where slickensides and "stairstep" breaks would allow determination of past movement. The largest of these zones cuts through a subparallel dike swarm in Blocks 26 through 29. Slickensided breaks in this zone were found in Monolith 28. They trended NE-SW, were near-vertical, and showed at least two episodes of movement in different directions.

Extensive, near horizontal fractures occurred within the rock excavated for the dam and powerhouse. Three open, leached fractures cutting the Georgia abutment strike perpendicular to the dam axis and dip gently toward the river, ending at the rock face between Blocks 7 and 8. Several subhorizontal fractures filled with decomposed rock were exposed during excavation of the spillway blocks. On the South Carolina abutment, several monoliths were excavated below design grade to remove low-angle fractures filled with clay. All of these fractures were subparallel to erosional and top-of-rock surfaces. They were interpreted to be relief joints.

The foundation rock is extensively jointed. Most of these joints are healed with epidote, chlorite, or calcite. Few open joints, except those opened by blasting, were seen below the zone of weathering. However, a few large cavities partially filled with euhedral pyrite, epidote, sintered quartz, and fine transparent needles of an unidentified mineral were seen in the intake and powerhouse areas. Individual joints in the foundation surface are noted on the geologic maps. Where a number of healed joints or veins surface in the foundation they are depicted as chlorite, epidote, or calcite streaks.

e. Bedrock Weathering: Weathering in the Richard B. Russell Project area, as in the rest of the southeastern United States, has been rapid, with thick accumulations of weathered materials between the weathering and erosion surfaces. Faults, shear zones, and open joints allow maximum penetration of meteoric water into rock masses, and mafic dikes provide more easily weathered rock, particularly if they are broken. Since these intersecting, near-vertical (in the dam area) and essentially planar features tend to weather more rapidly than the rock around them, a typical weathering surface in the area altlemates localized pockets of decomposed rock with pinnacles of relatively unweathered rock nearby. This was the type of weathering surface excavated on the South Carolina abutment.

With the exception of a large pocket in the upstream part of Blocks 2 and 3, the Georgia abutment had a relatively regular weathering surface dipping toward the river. This surface extended into Monolith 13, then began to flatten out under the river. The intake and spillway sections, which were mostly excavated into the riverbed, had a thin zone of weathering. In the most extreme case, it was about 10 feet from the top of

rock to fresh, unweathered rock. The tailrace area, also under the river, had a similar zone of weathering. However, in the part of the tailrace immediately behind the pumpback storage units, near-vertical, round-shouldered, open breaks in the rocks caused some problems. These joints were filled with alluvium, including large river-rounded cobbles.

Class I excavation on the excavation contract was greater than estimated (123% overrun). Most of this was removed from Blocks 1-14 and 25-32, with the majority of the overrun in South Carolina. This was, in large part, due to the fact that the exploratory holes in that abutment fortuitously penetrated the pinnacles, rather than the pockets of decomposed rock. The quantity estimated from this information was accordingly low.

f. Leaching and/or Solution Activity: Solution was not a major weathering factor in the metamorphic rocks at the damsite. Formerly calcite-healed veins were open at the surface, where water could dissolve the filling material, but there were no limestones occurring near the surface in the general area. All rock types which were broken to the point that meteoric or goundwater could penetrate them showed leaching. This was usually evident in that feldspars would be a pasty white color, and in some cases would be completely kaolinized. Leaching along joints and faults was of limited extent, being mostly confined to minerals immediately adjacent to the water-conducting break.

Chemical weathering processes such as hydrolysis and oxidation produce from primary minerals materials which can be dissolved or leached as finely divided suspensions or colloids. The silica hardpan, which can occassionally be seen in soil horizons in the general area, is formed from a colloidal product of the hydorlyzation of feldspars. Another product which was present and caused some trouble in excavation for the dam was kaolinite clay. This material, removed by groundwater from the rock mass as a finely divided suspension, helped form a slick or "fat" clay which settled in available open joints, including near horizontal relief joints. Since this material lowered frictional resistance to sliding, it was necessary to remove such joints. These "clay seams" were present in both abutments, but were not deep in the Georgia abutment.

A product of oxidation which caused minor trouble was iron salts. Pyrite and other iron sulfides are ubiquitous as accessory minerals in the foundation rocks. As these minerals were oxidized underground, soluble ferrous salts were produced and were leached by groundwater. When this groundwater surfaced in the excavation, the soluble salts were further oxidized into insoluble ferric salts. Some of the intake and spillway blocks were excavated for over a year prior to placement of the last lift of concrete on rock. Where groundwater ran over this rock, ferric salts adhered to the surface. This material proved very difficult to remove, even with a water-blaster.

g. Groundwater: Water table measurements in exploratory core holes prior to excavation showed that the top of groundwater in the right (Georgia) abutment varied from a few feet to 70 feet below top of ground. However, the overall piezometric surface was reported to be very gentle. On the left (South Carolina) abutment, the gradient was much steeper close to the river bluff. However, further from the river it also assumed a very

gentle gradient. This was attributed to possible faulting providing easy drainage near the river. Excavation later proved out this observation.

- 3.3 Geologic Structure: The geologic structure of the damsite will be discussed under three subtopics: (1) Folds, (2) Joints, and (3) Faults.
- 3.3.1 Folds: Within the damsite folding on the megascopic scale consists of a northeast striking, steeply dipping to the southeast, isoclinal fold. The dip averages more than 55 degrees. This fold is regional in scale and is likely related to the accretion onto the North American craton of the Avalon terrane.
- 3.3.2 <u>Jointing</u>: Jointing in the rock at the damsite assumes two main styles. They are: (1) near vertical, intersecting joints, and (2) exfoliation, or relief joints which are sub-parallel to the topographic surface. These two styles will be discussed in turn in the following paragraphs.

The most consistent and persistent joints at the damsite are an intersecting set of near-vertical joints with average attitudes of strike N46°W, dip 89°NE and strike N60°E, dip 89°NE (Design Memorandum 8 - Geology, 1974). This pattern results in the compressive component (Sigma 1) being resolved into a strike of N84°W and the extensional component (Sigma 3) striking N06°E. This stress orientation is consistent with that which the Design Earthquake Report (COE, 1977) determined for the region. Also, the strike of Sigma 3 is consistent with the hypothesis that the Diversion Channel Fault possesses a strike-slip component. Table 2 and Figures 5 thru 8 summarize joint orientations measured during design phase. Locations where measurements were taken are shown on Plate 5.

The majority of the joints within this set have low-asperity faces and are tight. Many have also been healed with calcite, and iron sulfides are also fairly common on the joint faces. The average spacing for these joints is about one foot (30 cm).

The second type of jointing to be discussed is exfoliation or stress relief jointing. These are large-scale structures, sub-parallel to the topographic surface, which were formed by dilation of the rock mass following removal of the overlying confining pressure by erosion. These joints are closer-spaced near the surface than at depth, due to the fact that less residual stress is required to initiate fracturing under conditions of decreased overburden stress. Similar joints were created by excavation for the concrete dam and powerhouse, though on a much smaller scale. Stresses imposed by blasting operations, together with existing stresses, were often of such magnitude as to create joints in some instances, and in others, the combined stresses could only create incipient joints which, when subjected to the added stresses imposed by mechanical excavation equipment during foundation preparation, resulted in a seemingly endless repetition of removing rock from the same area time and again.

3.3.3 <u>Faulting</u>: The major fault within the damsite crosses the Georgia embankment in the vicinity of the Stage I diversion channel and is referred to in this report, and most other C.O.E. reports on this project, as the Diversion Channel Fault. Plates 58 and 60 show the location of the fault

TABLE 2
ORIENTATION OF FRACTURE PLANES

Location*	No. of measurements	Fracture planes Strike Dip
1,2	301	1. N37W 79NE 2. N60E 84NW
4	47	1. N43W 77SW 2. N56E 87SE
5	141	1. N35W 83NE 2. N83W 82SW 3. N60E 79NW
6	55	1. N38W 88NE 2. N62E 88SE 3. N79W 79SW
7	62	1. N42W 85NE 2. N80W 74SW 3. N74E 81NW
8	61	1. N42W 86NE
9	184	1. N37W 86SW 2. N66W 82SW 3. N54E 78NW
10	85	1. N38W 88SW 2. N53E 84NW 3. N73W 87SW
13	220	1. N34W 79NE 2. N54E 88SE 3. N84W 83SW

^{*} Refer to Plate 5 for locations

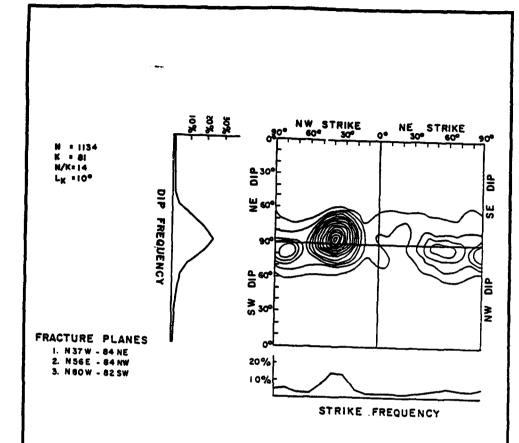
TABLE 2 (CONTINUED)

Location	No. of measurements	Fracture planes Strike Dip
14	234	1. N40W 90 2. N59E 86NW
15	265	1. N46W 89NE 2. N60E 86NW
16	56	1. N45W 86NE 2. N60E 90

^{*} Refer to Plate 5 for Locations

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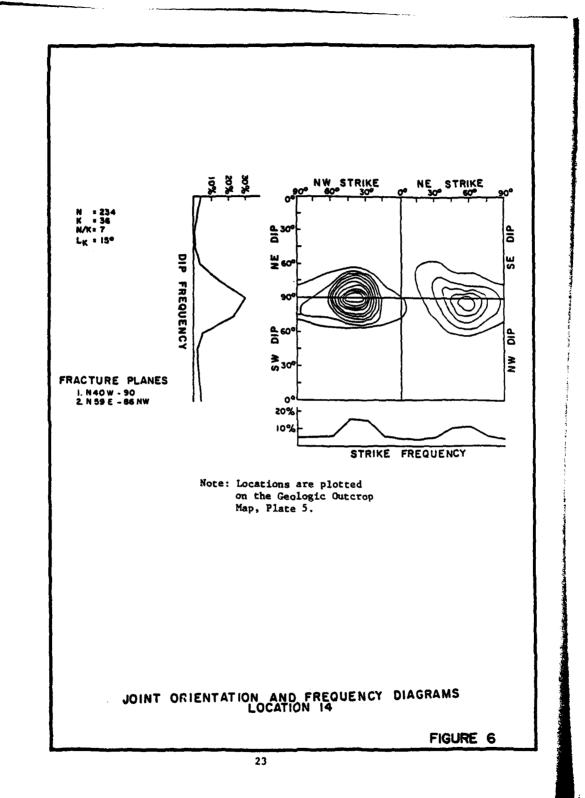
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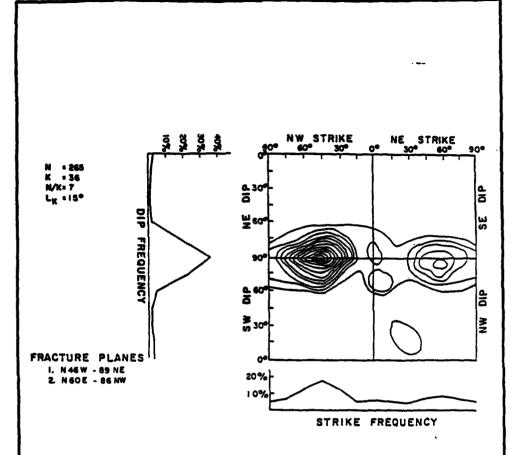


Note: Locations are plotted on the Geologic Outcrop Map, Plate 5.

JOINT ORIENTATION AND FREQUENCY DIAGRAMS LOCATIONS 1, 2, 5, 6, 7, 8, 9, 10 & 13 COMBINED

FIGURE 5

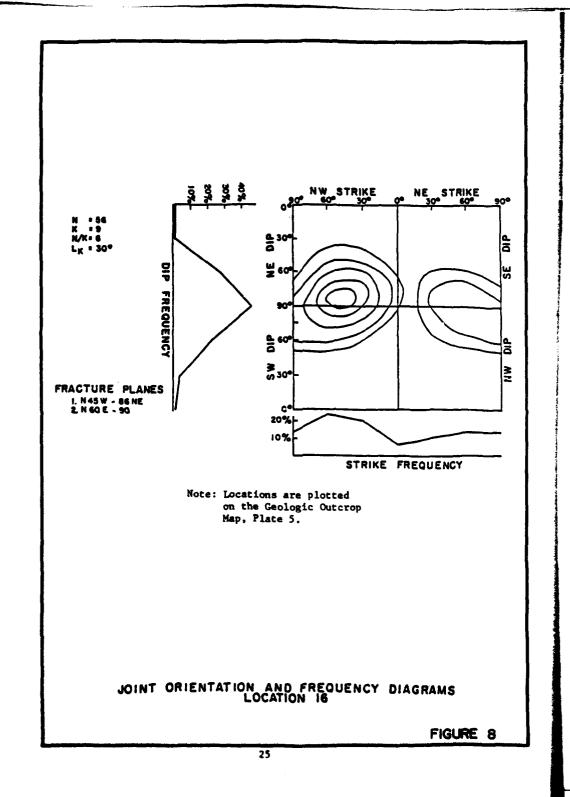




Note: Locations are plotted on the Geologic Outcrop Map, Plate 5.

JOINT ORIENTATION AND FREQUENCY DIAGRAMS LOCATION 15

FIGURE 7



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relative to the major features of the project. The existence of this fault was suspected during the exploration phase of the project, but was not confirmed until excavation for the diversion channel exposed it.

The Diversion Channel Fault strikes approximately N10°E and the dipvaries from about 45°NN to 55°NN. Indicators of both dip-slip and strikeslip movement are present. These are slickensides, offset lithologic units, truncated lithologic units, and differential weathering across the fault which lead to the hypothesis that the fault has been activated a number of times. The most convincing evidence of this is the presence of a soft gouge zone within the felsite which has intruded into the fault.

The origin of the fault has not been clearly determined, but as Wheeler and Bollinger (1984) point out, "the narrow ends of the Avalon terrane are its only seismically active parts, and are the most likely parts to have had fractures initiated and/or reactivated during the growth, transport, and accretion of the terrane".

Faulting of the Avalon terrane could be the result of reactivated North American cratonic or Ispeton faults. The Diversion Channel Fault may, however, be syngenetic with the Elberton or Denburg batholiths. However, if older than these intrusives, it was almost certainly reactivated by then. Evidence for this line of reasoning is the existence of the felsite filling of the fault and the hydrothermally-altered zone branching off the fault along the upstream limit of the cutoff trench (Plate 77).

3.4 Engineering Characteristics of Overburden Soils:

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- a. Residual: Residual soils generally blanket the area except in the Savannah Riverbed and where rocks outcrop. The thickness of these soils varies from a few feet to over 50 feet. Residual soils consist of mixtures of sand, silt, and clay classified as SM, SC, ML, CL and CH (Unified Soil Classification System). The soil blanket is generally thin over much of the area. This caused a great deal of difficulty in finding adequate sources of borrow for earthfill.
- b. Alluvial: Alluvial soils occur in the Savannah Riverbed as mixtures of sand, silt, and organic material (SP, SM and OL classifications). At the dammite, they are about 25 feet thick. The SP and SM soils are predominant.

A detailed testing program of all major overburden materials is outlined at length in Design Memorandum 17 - Earth Embankments and the Richard B. Russell Earth Embankment Criteria and Performance Report. The testing was done largely in consideration of usage as borrow material, dynamic stability, and in-situ characteristics.

3.5 Engineering Cheracteristics of the Bedrock Materials: Following is a compilation of information about bedrock material which was taken from Design Memorandum 8 - Geology .

3.5.1 Physical Properties of the Foundation Rock:

a. Prophyritic meta-dacite and associated basic intrusives underly the proposed dam. Field observations and laboratory tests indicate the sound rock to be strong and computent. Physical tests were made on representative samples of porphyritic meta-decite and meta-diabase. Cross-hole seismic velocities were measured at verious elevations in sound rock under a large portion of the concrete dam foundation, and several up-hole seismic velocity measurements were made for comparison with laboratory sonic velocities. Table 3 is a summery of the physical tests; Table 4 lists cross-hole seismic velocities; Table 5 lists up-hole seismic velocities, and Table 6 lists the laboratory sonic velocities and average Schmidt hammer rebound numbers from intact samples of rock core. Figure 9 is a plot showing the relationship of sonic velocities and Schmidt hammer rebound numbers.

- b. Unconfined Compression Tests. An average value of 24,280 psi was obtained from the unconfined compression tests performed on three representative samples of meta-diabase and four representative samples of meta-decite. The average value of the meta-diabase samples was 20,150 psi and of the meta-dacite samples was 27,390 psi. These values reveal both rock types to be in the high strength classification. An allowable bearing value of 72 tons per square foot will be used in design.
- c. Triaxial Tests. Seven samples were subjected to triaxial tests using confining pressures of 1,500 psi and 3,000 psi. The internal friction angles varied from 23 to 58 degrees and the shear stress intercepts (cohesion) varied from 2,000 psi to 8,500 psi (see Table 3). These results further substantiate the high strength of the rock.
- d. Shear Strength and Coefficient of Sliding Friction. The direct shear strength values of the samples tested ranged from 2,115 psi to 4,075 psi. Coefficient of sliding friction tests were made on joint surfaces in the rock core. The coefficients of sliding friction varied from 0.98 to 3.61 (see Table 3) and averaged 1.96. An allowable value of 1.0 for the coefficient of sliding friction will be used in design. This value is considered conservative since it is essentially equal to the lowest of seven test values and approximately half the average. The allowable unit shear strength for design purposes is 500 psi. This value was obtained by using the lowest test value and also allowing for low angle joints over 75 percent of an assumed failure plane beneath the surface.
- e. Specific Gravity. The average specific gravity was 2.71 for the purphyritic meta-dacite and was 3.00 for the meta-diabase.
- f. Modulus of Elasticity. The average value of the static (initial tangent) modulus of elasticity was 7,630,000 psi. The average of three values of the dynamic modulus (Table 3) was 5,700.00. Additional dynamic values were measured, but the results were questionable (see Table 3).
- g. Velocity Measurements. Tables 4, 5, and 6 show the results of cross-hole, up-hole, and laboratory sonic velocity measurements. The 44 cross-hole measurements ranged from 12,800 to 16,000 feet per second. The laboratory sonic velocity measurements ranged from 12,000 to 19,600 feet per second. All measurements were of sound rock velocities or between sound rock zones in adjacent drill holes. Figure 9 shows the excellent correlation between the average Schmidt hammer rebound numbers and sonic velocities. The sonic velocity measurements were all made on dry samples

TABLE 3

SUMMARY OF PHYSICAL TESTS

Lab no. rock type	Hole no. sample depth	Sulk specific gravity	Unconfined compressive strength PSI	Poiss- ons ratio	Tong 10 and 15 a	Modulus of slasticity PS	Modulus of stricity PSI x 10 ⁶	Direct sheet strength-PS	shear th-PSI	Coefficial last frical last design	ion (slid- normal load)	Jaconal Internal friction	Cobes ion
						Static	Dynamic	Rock	Concrete on rock	pck rock	Cancrete on rock	degroes	
IM 2512 meta-diabase	C-146 21'	2.2	23,550	0.29	2,000	7.14	0.54	3,590	440	1.67	1.2	*	÷,68
IN 2513 C-130 qtz-fojd-gneiss 28°	C-130 iss 28°	2.78	26,720	0.24	10,000	5.20	5.78	3,250	700	1.51	1.15	ถ	8,300
IN 2514 meta-diabase	c-7 3\$'	3.6	10,960	0.21	13,000	12.90	0.40	4,075	231	1.49	•	a	2,000
IN 2515 qtz-feld-gneiss	C-129 1ss 15'	2.70	31,940	0.30	4,000	5.20	7.43	2,180	165	3.61	į	*	9,500
IM 2516 meta-diabase	C-29 34°	3.02	25,930	0.25	9,000	13.79	1.05	3,190	8	1.57	į	\$	2,000
IN 2517 qtz-feld-gneiss	C-132 Iss 30'	2.70	34,350	0.25	12,000	3.18	3.8	2,390	365	3.08	;	7	7,300
IM 2516 qtz-feld-gneiss	C-144 188 38'	1.1	16,540	6.21	20,000	5.97	0.24	2,115	134	0.8	1.8	:	:

Resaris:
*Sample contained healed fractures which probably interferred with frequency readings.
**Indeterminate, probably due to the variable strength of the steeply dipping fractures present in the rock.

TABLE 4

CROSS-HOLE SEISMIC VELOCITIES

Shot location hole no.	Shot elevation	Hydrophone location hole no.	Hydrophone elevation	Velocity fps
C-127	327.0	C-148	327.0	11,300
C-127	327.0	C-146	327.0	13,400
C-127	327.0	C-147	327.0 327.0	12,800 14,500
C-127	327.0	C-145 C-148	315.0	14,200
C-127	315.0	C-146	315.0	13,400
C-127	315.0 315.0	C-147	315.0	17,100
C-127 C-127	315.0	C-145	315.0	17,000
C-130	300.0	Č-129	300.0	15,000
C-130	300.0	C-131	300.0	13,800
C-130	290.0	C-129	290.0	17,600
C-130	290.0	C-131	290.0	16,900
C-131	300.0	C-129	300.0	16,700
C-131	300.0	C-132	300.0	12,200
C-131	290.0	C-129	290.0	18,100 16,800
C-131	290.0	C-132	290.0 289.5	11,000
C-130	303.5	C-151 C-152	285.0	11,500
C-130 C-130	303.5 303.5	C-151	299.5	10,500
C-130	303.5	C-152	295.0	11,200
C-152	285.0	C-151	285.0	10,200
C-152	285.0	C-153	285.0	9,500
C-152	295.0	C-151	295.0	10,000
C-152	295.0	C-153	295.0	9,800
C-152	288.5	C-151	288.5	10,000 10,000
C-152	288.5	C-153	288.5 290.4	10,000
C-134	293.4	C-164	290.4	9,500
C-134	293.4 293.4	C-135 C-164	280.4	16,500
C-134 C-134	293.4 293.4	C-135	280.4	14,800
C-134	286.2	C-137	286.2	10,700
C-136	286.2	C-135	286.2	10,100
C-136	276.2	C-137	276.2	10,400
C-136	276.2	C-135	276.2	9,800
C-155	278.2	C-137	278.2	11,300
C-155	278.2	C-138	278.2	12,700 11,500
C-155	288.2	C-137	288.2 288.2	12,500
C-155	288.2	C-138 C-153	275.0	13,900
C-164	275.0	C-153	250.0	18,400
C-164 C-154	250.0 284.1	C-164	284.1	14,400
C-154	284.1	C-153	284.1	14,300
C-154	294.1	C-164	294.1	9,200
C-154	294.1	C-153	294.1	9,100

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TABLE 5

UP-HOLE SEISMIC VELOCITIES

Hole no.	Shot elevation	Hydrophone elevation	Velocity
C-153	250.0	300.0	12,800
C-164	240.0	290.0	16,100
C-152	247.0	297.0	14,700
C-151	240.0	290.0	15,600

TABLE 6

SONIC VELOCITIES AND AVERAGE SCHMIDT REBOUND NUMBERS

Hole no.	Elev.	Rock_type	Rebound no.	Velocity
C-127	315	Meta-diabase	51	19,600
C-129	292	Otz-feld-Gneiss	39	12,000
C-130	292	Meta-diabase	54	21,600
C-146	314	Meta-diabase	52	18,000
C-131	290	Qtz-feld-Gneiss	50	18,800
•		Flaser Texture		40,000
C-132	292	Otz-feld-Gneiss	40	12,100
C-152	285	Otz-feld-Gneiss	50	16,600
J		Flaser Texture	• •	20,000
C-153	286	Otz-feld-Gneiss	42	12,500
C-154	280	Otz-feld-Gneiss	46	15,200
C-164	271	Otz-feld-Gneiss	46	15,100
		Flaser Texture	. •	,

TABLE 7

CORRELATION OF RQD AND SEISMIC VELOCITIES

Description of rock quality	RQD percent	Seismic velocities (fps)
Very poor	0-25	6,000 - 8,500
Poor	25-50	8,500 - 11,000
Fair	50-75	11.000 - 13.500
Good	75-90	13.500 - 15.000
Excellent	90-100	15,000 -

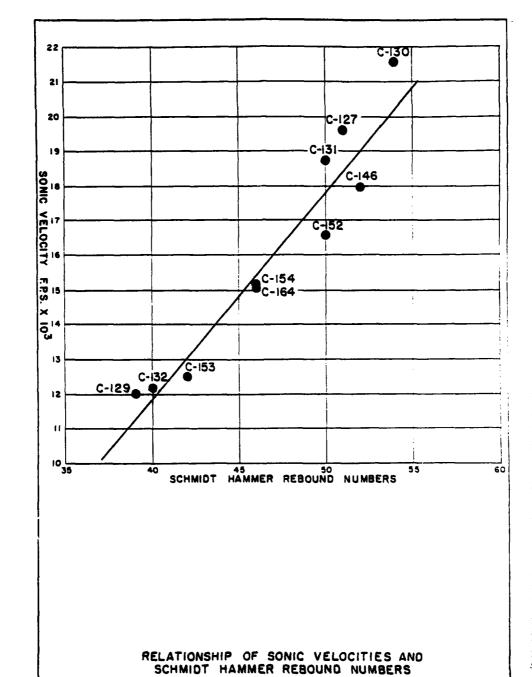


FIGURE 9

with no pressure applied. Tables 4-6 tabulate the lowest elevation cross-hole velocity, up-hole velocity, and sonic velocity measurements and elevations. The following conclusions are the result of velocity studies to date at the Richard B. Russell project:

- (1) Velocities are greater in the meta-diabase (specific gravity 3.0) than in the meta-dacite (specific gravity 2.71). Laboratory sonic velocities of three meta-diabase samples varied from 18,000 to 21,600 feet per second. Of seven meta-dacite samples, only one had a sonic velocity above 17,000 feet per second, and the average was 14,600 feet per second.
- (2) Alteration of feldspars to kaolinite (less than 10 percent kaolinite) apparently causes a substantial reduction in sonic velocities, but not in unconfined compressive strengths. Evidence for this statement is based on tests performed on the rock from hole No. C-129. Petrographic analysis, physical tests, and sonic velocity measurements were all performed on samples from this hole. The kaolinite content was approximately 8 percent and the sonic velocity 12,000 feet per second (the lowest value of 8 samples); however, the unconfined compressive strength was 31,940 psi (the second highest of seven tests).
- (3) For velocity ratio studies, sonic velocities probably should be measured under saturated conditions using 3,000 to 5,000 psi uniaxial applied stress. Laboratory velocity (VL) measurements on sound intact samples of rock normally give higher velocities than do field seismic (VF) measurements). The velocity ratio, VF/VL, has been proposed as a method of evaluating the degree of jointing and fracturing in the rock. Theoretically, the difference between the two values is caused by discontinuities in the rock. In a massive, tight, fresh rock mass, the velocity ratio should approach unity. As discontinuities increase, the velocity ratio should get smaller. At this site, the theory appears to be valid in the meta-diabase rock, but in the more predominant meta-dacite rock, it does not hold up. In almost every case, the laboratory sonic velocity of intact specimens of meta-dacite was less than the field seismic velocity in the same boring. This would give velocity ratios greater than unity. Up-hole seismic velocities were measured in the field to see if direction of measurement was causing this discrepancy. The results did not indicate this to be the answer. It was finally concluded that the conditions of in-situ stress and saturation of the rock in the ground must be at least a partial explanation of the failure of the velocity ratio method in this case.
- (4) Additional subsurface investigations must be made to determine why the cross-hole velocities were so slow (10,000 fps) between holes C-135, C-136 and C-137, C-151 and C-152, and C-152 and C-153. Deep weathering or faulting between these holes could explain the abnormally low velocities.
 - (5) Velocities increase with depth.
- (6) Comparison of seismic velocities with the rock quality designator (RQD) is more meaningful at this site than a comparison of the velocity ratio with the RQD. Table 7, on page 30 shows approximate, assumed correlations between seismic velocities (including refraction, cross-hole, and up-hole) and the RQD for the Richard B. Russell area. Enough

measurements have been made to indicate a good degree of confidence in the correlations.

4.0-CD SPECIAL DESIGN CONSIDERATIONS - CONCRETE DAM

4.1-CD Uplift Forces: Uplift on the base of the concrete dam was assumed to vary according to the following criteria: full tailwater at the toe of the dam to tailwater plus one-half the difference between headwater and tailwater at the heel of the dam. Uplift pressure is considered to be effective over the entire base of the dam. When examining the stresses at any horizontal plane through the structure above the base, the uplift pressure is assumed to be 50 percent effective over the entire area of the plane under consideration.

To contend with uplift, a system of gallery drains was drilled into the foundation rock. Angled 15 degrees in the downstream direction, the drain holes were drilled to an average depth of 80 feet into the foundation rock beneath the dam. These holes were cased within the concrete structure, allowing several inches of "stick-up" above the gutter elevation. The tops of each drain allow water to flow out of the drains and into the gallery gutter, wherein it is led to one of two sump pumps for final elimination from the inspection gallery. (See gallery drain details on Plate 50.)

In order to effectively measure uplift pressure, a pattern of uplift cells was designed and installed for measuring and reading pressures. The cells permit water to enter through well screens and act against pressure gauges. The pressure gauges are located conveniently in the inspection gallery where they may be read individually by monoliths. For details of uplift cells, refer to Plates 53, 54, and 57.

4.2-CD <u>Foundation Bearing</u>: In design, an allowable bearing of 72 tons per square foot (or 1000 psi) was established. In the testing of the seven rock samples previously mentioned under Section 3.5.1 average strengths of 27,390 psi for meta-dacite and 20,150 psi for meta-diabase were determined. The average static modulus of elasticity for all samples was 7,630,000 psi. The lowest unconfined compression strength result of 10,960 psi still yields a safety factor of 10.96 or nearly 11.

4.0-EE SPECIAL DESIGN CONSIDERATIONS - EARTH EMBANKMENTS

4.1-EE Overlap Grout Holes: In order to obtain effective grout coverage in the regions of the Georgia earth embankment where it abuts the concrete dam structure, a series of angled fan grout holes was designed. Drilling and grouting this pattern of grout holes permitted a consolidation of the embankment grout curtain and the grout curtain beneath the concrete dam. This fan arrangement is best understood by referencing Plates 86 thru 92.

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4.2-EE Fault Treatment: The fault uncovered during diversion channel excavation proved to be an item for modest concern during embankment design. Thought to be, for all purposes relatively inactive, its large size dictated some special design consideration. (Water losses during pressure testing had been large in the vicinity of the fault also). A blanket or consolidation type grout program was designed for intercepting the fault plane at various elevations determined by the plane of its orientation. In addition, a surface grout cap was included in the contract drawings for adequately covering the excavated surface along the grout curtain from which

the consolidation holes would be drilled and grouted. Details of the consolidation grouting, as well as the concrete cap, can be found on Plates 82, 108, and 109. The results of the consolidation grouting program, which was field-modified somewhat, are discussed in Section 7.1.4-FE.

4.0 PH SPECIAL DESIGN CONSIDERATIONS - POWERHOUSE

4.1-PH Hydrostatic Uplift: The powerhouse structure was designed to utilize gravity design, an extensive rock bolt network, and foundation relief wells to counteract hydrostatic uplift forces. Section 8.0-PH describes the rock bolts and anchors utilized in the structure, the quantity and locations, and the engineering rationals for their employment. Section 10.0-PH covers the relief well system utilized. Design studies concluded that the combination of elements which were used would be preferable to a totally gravity design. This conclusion was based upon both economic and constructabilty considerations.

4.2-PH Spillway Training Wall: The spillway training wall, which extends downstream from the South Carolina side of Unit 8, not only has to carry the vertical loading imposed by the observation tower, but also has to resist rotational and overturning forces imposed by tailrace and spillway water currents. To meet the stability criteria, a series of twelve (12) twenty-three strand, high-capacity Type "C" rock anchors (Plates 127 and 129) and a series of sub-horizontal Type "A" rock bolts (Plate 128) were incorporated into the design. In addition to this, the designers requested that the wall be "keyed" into the rock and the Area Office staff added five (5) twenty-foot long Type "D" rock bolts which were installed vertically through the second lift of concrete and grouted into the rock with polyester resin. After the grout set up, the locking nuts on the bolts were torqued with a manually operated wrench rather than stressing with a hydraulic jack as called for by the specifications. These Type "D" bolts were used primarily to maintain concrete-to-rock contact and to guard against the possibility of Unit 8 discharge water exerting any uplift forces against the bottom of the wall.

5.0-CD EXCAVATION PROCEDURES FOR COMPONENT PARTS - CONCRETE DAM

5.1-CD Excavation Requirements: The excavation phase of dam and powerhouse construction was undertaken as a separate contract to expedite completion of the project. The point of separation was that the excavation contractor would take the excavation to design grade, clean up by mechanical means only (with some washing of the surface provided for in the contract), perform needed dental excavation, drill exploratory holes as needed to confirm the suitability of the foundation rock, then demobilize. The concrete dam contractor would perform foundation preparation by removing a minor amount of rock by hand methods, wash the entire foundation, then construct the dam on the resulting foundation.

The excavation contract defined two classes of excavation. Class I was that material which could be removed by diligent use of a single shank hydraulic ripper on a D8H bulldozer. Class II was that material which had to be systematically drilled and blasted. Within Class II, "sound rock" was further identified as rock that is hard, intact, fresh, and essentially unweathered.

Slopes were set based on the material being excavated. Class I excavation slopes were variable, but in no case were they allowed to be steeper than IV on 1.5H. Firm rock (Class II rock not classified as sound) slopes were to be 1 on 1. Sound rock slopes were vertical. A bench or berm not less than 6 feet wide was to be left between firm and sound rock slopes.

5.2-CD <u>Excavation Grades</u>: Excavation grades as designed and directed in the field are depicted on Plate 16. As-built foundation grades after foundation rock removal and preparation are depicted as topographic surfaces on Plates 18 through 49.

Directed changes occurred in 3 main areas. The first of these was the Georgia abutment. After Class II excavation reached design grade in Monoliths 2 and 3, a large pocket of decomposed material appeared to be both weathered and hydrothermally altered. It was easily removed by a Caterpillar 245 backhoe.

The second area is the spillway foundation. In this area (probable) relief joints in a near horizontal attitude occurred below design grade. The first of these was discovered when communication was noted between holes RBR10A and RBR18 in Monolith 19. Material filling the seam was decomposed and crushed rock fragments of sand and gravel size. Both metadacite and metadabase were fragmented where intersected by the seam, and lateral displacement of near-vertical dikes along the seam was 0 to 6 inches.

The third area was in the South Carolina abutment. Monoliths 26, 27, and 28 were founded at lower elevations to remove clay seams below design grade. These seams were also near horizontal and were probably relief joints. Exploration drill holes indicated that these seams probably terminated at their intersection with metadiabase dikes. Monoliths 27 and 28 were lowered by modification after the excavation contractor had essentailly completed his contract work.

Minor excavation had to be done on the concrete dam contract. The dewatering sump in the Block 9 inspection gallery was designed to be founded below the excavation surface left by the Lane Company. In order to avoid interference between contractors, the Eravo-Groves contract was modified to allow them to blast and remove the unwanted rock.

5.3-CD <u>Dewatering Provisions</u>: When excavation was begun in 1977, the construction area lay between two cofferdikes approximately 1000 feet upstream and downstream of the concrete dam. A sedimentation dike, constructed of sand and gravel, created a triangular pond immediately downstream of the downstream cofferdike. The water level in this pond responded to water level changes in Clarks Hill Lake. The spillway and part of the intake areas were low, swampy, wandering watercourses immediately after unwatering.

The excavation contractor dug a downstream sump and then built an elevated (on random fill) road immediately upstream of the construction area. Water which pended upstream of this road was led across the Monolith 22 location by a ditch with piled fill on either side to the downstream sump. From the downstream sump, the water was pumped over the downstream cofferdike into the sedimentation pend. Since the sedimentation dike was permeable to water, but not to silt and clay, the water filtered into Clarks Hill Lake, leaving the sediment behind during periods in which Clarks Hill Lake, leaving the sediment behind during periods in which Clarks Hill Lake was low. When the lake was high, water filtered in reverse flow through the dike cleaning the accumulated silt and clay from within the sand and gravel fill.

When excavation was proceeding at its fastest pace in the tailrace and powerhouse areas, two 8-inch pumps were required to empty the excavation into the downstream sump. During this period, work on the concrete dam was going on, adding more water to the amount of dewatering necessary. A large pump of unknown capacity would then move this water from the sump into the sedimentation pond. This plant was capable of keeping up with runoff and groundwater flows in all but the heaviest thunderstorms. Exact quantities of water pumped were not kept, since the contractor was paid by the month instead of by a unit quantity. Appendix B was compiled from contractor records available for a 6-month period in 1978.

5.4-CD Overburden Excavation: Overburden excavation began with the dredging of river sand prior to Sewatering. The dredged sand was moved as a slurry to the sand stockpile by pupeline with the help of two booster pumps. When dredged material was considered by Corps inspectors to be sufficiently contaminated by muck, it was sent to the waste pile. By contract, the riverbed was dredged to within 2 feet of rock.

After dewatering, Class I excavation was accomplished with bulldozers (Caterpillar D8's, end loaders (Caterpillar 988's), and off-road dumps (Caterpillar 769's). The dozers broke up and piled the material, the loaders filled the dumps with it, and the dumps hauled it to the stockpile, or more commonly to the waste pile. On the abutments, the Class I excavation material was pushed downhill or over the bluff where it was loaded. There was a contract overrun-of 123% in Class I excavation in the concrete dam and powerhouse areas.

5.5-CD Rock Escavation: The contractor began drilling and blasting operations in November 1977, on the South Carolina abutment. He began drilling on the Georgia side of the excavation in January 1978. After that time, he always worked at least two areas at the same time. At the beginning of operations, the area from the downstream portion of Monoliths 8 through 16 was held up by the Corps (pending completion of design for the powerhouse) to allow no excavation below 277 MSL. This prohibition was relaxed in April 1978.

The contractor drilled his blast holes with air-trac drills, working as many as six at one time. Although the contract specifications allowed holes up to 6 inches in diameter, the contractor elected to use 3 and 3-1/2-inch holes on his production blasts. Blasts to grade were limited by contract to 2-1/2-inch diameter holes. Maximum lifts as shot were 16 to 20 feet. Although a maximum lift of 5 feet for shots to grade was specified, when there were only 7 or 8 feet of rock to be excavated below ripper refusal this was shot in one lift. The most common pattern used in production blasts was a 5-foot by 6-foot rectangular pattern. This pattern was closed up for blasts to grade.

The production blasts, as designed, generally achieved good breakage, but backbreak was intensified by the jointed nature of the metamorphic rock. Healed and incipient joints were opened by high pressure gasses. Calcite sheets could often be easily lifted from formerly healed joints now opened by blasting in the deeper powerhouse shots. It was difficult to hold backbreak to less than 6 or 7 feet in production blasts. This tendency of the rock to split along pre-existing joints particularly affected the slopes in the downstream portions of Blocks 6 through 16. After foundation preparation "stairstep" slopes were common in these blocks.

Another difficulty encountered in drilling and blasting operations was the problem of caving holes. As previously stated, the weathering surface in the South Carolina area was one of pinnacles and pockets. Where hard rock surrounded and protected pockets or crevices filled with decomposed rock materials, penetrating blast holes often caved repeatedly. These pockets and crevices affected blasts for 2 or 3 consecutive lifts in some cases.

The Lane Company used DuPont explosives in production blasts. Holes were loaded with ANFO prills and detonated by electric caps in HDP primers where possible. Where water in the holes or oversized holes dictated, Tovex 700 certridges of 2 and 2-1/2-inch diameter initiated with EM caps in Detaprime UA primers were used. Blasts were usually delayed in a chevron pattern to give a humped muck pile which facilitated loading with a front end loader. In the harder rock deep in the powerhouse excavation, a rectangular shot would sometimes be delayed in rows toward one corner (where two free faces existed) to get better breakage. Powder factors were usually one pound per cubic yard or less in firm rock. These ranged up to two or two-1/2 in the powerhouse trench. Blast No. BP1205, a sinking shot in the powerhouse area with holes angled toward and delayed to the center of the shot, had a factor of three.

Shot rock was loaded onto Caterpillar 769 off-road dumps by Caterpillar 988 loaders assisted by D8 bulldozers. As the rock was loaded, Corps

inspectors designated it as firm or select rock. The rock was then taken to the appropriate firm rock or select rock stockpile. All the rock removed from the dem and powerhouse excavation was used in construction. Firm rock was primarily used to build construction roads. Select rock was used as filter material and rock fill in the earth embandments.

5.6-CD Line Drilling and Presplitting: The excevation contract specified line drilling (specing on holes not to exceed twice the diameter) "whenever a required vertical surface is less than four feet in height (depth); and for a horizontal distance of five feet on either side of all external corners for the full depth of the vertical surfaces." Presplit blasting was required on all surfaces inclined IV on IH or steeper which were 4 feet or greater in height and which were not required to be line drilled. Spacing on presplit holes was to be 20 to 36 inches.

Presplit holes were loaded with Hercules Fowder Co. Hercosplit WR. Most blasts were loaded solid with these 2-foot long (.6 lb/foot) cartridges, but in some cases, the holes were deck loaded with cartridges spread a foot or more apart. The Hercosplit is a nitroglycerin-based explosive and was initiated with primacord. Tovex 1, a water gel explosive in 50-foot long, 1-inch diameter, plastic tubes with internal primacord was tried on the South Carolina abutment, but the Contractor discontinued its use after one blast.

5.0-EE EARTH EMBANDENT - EXCAVATION

- 5.1-EE Excavation Grades: The excavation grades for the embankments are shown on Plates 60, 61, and 62. These plates show both the design grades and the "as-built" grades of the cutoff trenches. For more details, see the Embankment Performance and Criteria Report for this project.
- 5.2-EE Dewetering: Contract provisions for dewatering of the embankment considered only the diversion channel between the closure dikes. Design grades for the cutoff trenches assumed that "slope to drain" configuration could be achieved and detailed dewatering provisions were not written into the contract. This turned out not to be the case for all sections of the cutoff trenches and this condition required pumping and/or the blasting of drainage trenches through high points of the foundation rock. The area between Station 5+50 and Station 6+50 (referred to as the "bathtub" and whose treatment is presented in detail in Section 7.3-EE) presented the greatest problems for dewatering.

Dewatering of the diversion channel between the closure dikes required pumping from the time of initial closure until fill placement began (a period of eight months duration). The dewatering system consisted of one eight—inch submarsible pump which proved to be effective except on two occasions when excessive seepage through the upstream closure dike flooded the cutoff trench. Details concerning the closure dike seepage are presented in the Embandment Performance Report.

5.3-EE Overburden Excavation: Excavation of overburden for the cutoff trenches included alluvium, residual soils, and intensely weathered rock. Overburden excavation was classified as that material which could be removed by equipment such as scrapers, dozers, rippers, and hydraulic excavators

without resorting to drilling and blasting and was paid for under Bid Item 20: Excevation, Common. Ripper refusal criteria was based upon the use of a single-shank, perallelogram, hydraulic ripper mounted on a Caterpillar tractor Model DSH, 46A (or equivalent). All cutoff tranches were excevated to ripper refusal (as a minimum) except in the transition slopes, the "bathtub", and the diversion channel crossing. All material which was excevated was classified by a Government Inspector and was either stockpiled for future use or was wasted, based upon their classification.

The side slopes of the excavation were sloped to a IV on 1.5H and presented no stability problems. The contractor was required, however, to maintain the slopes by constantly monitoring and scaling, and to provide whatever support might be required.

5.4-EE Rock Excavation: The earth embankments cutoff trench was excavated to ripper refusal using a Caterpillar D8H 48A, equipped with a single-shank parallelogram ripper. Excavation below this level was performed only in isolated instances for the following purposes: (1) to remove high "knobs" of hard rock; (2) to provide drainage; and (3) to provide a smooth transition into the diversion channel. All rock excavation for the preceding purposes was performed using explosives. A total of approximately 47,500 CY of rock was excavated under the embankments contract, with the vast majority of this quantity being required to achieve a smooth transition into and across the diversion channel (Plate 82). The only other area with a substantial quantity of rock excavation was in the South Carolina abutment, where approximately 7,500 CY of rock was blasted to achieve a smooth transition. All other blasting consisted of small pop shots to remove small knobs or ledges of rock to achieve drainage or to provide for more efficient construction of the embankment.

Blast patterns varied from 4x4' to 5x7' to individual holes. The powder factor was on the order of 1.1 lb/yd and gelatin dynamite was the most commonly used explosive. All shots were detonated with millisecond delay electric blasting caps. Subdrilling was normally held to a maximum of 1 foot with a minimum of 3 feet of stemming. Lightning detectors and seismographs were required for all blasting operations.

5.0-PH EXCAVATION PROCEDURES FOR COMPONENT PARTS - POWERHOUSE

C

- 5.1-PH General: Excavation for the powerhouse structure was performed under a separate contract by the Lane Corporation of Meridian, Connecticut. Procedures, methods, and equipment are presented in detail in Section 5.0-CD. The only excavation performed under the Powerhouse Contract (other than foundation preparation and switchyard excavation) was that required for sumps in the erection bay and areas which were not shot to grade under the Excavation Contract.
- 5.2-PH <u>Erection Bay Sumps</u>: In the erection bay area, three (3) sumps had to be excavated in the foundation rock. This work was accomplished through the use of jack hammers, prybars, and Bristar. Bristar is a commercially available demolition agent which is non-explosive and which fractures the rock because of its expansive properties.

- 5.3-PH Penstock Drain Line: The penstock drain line for Service Bay 2 required blasting to remove high areas of rock. This blasting was performed by Piedmont Explosives, Inc. with vibration monitoring and evaluation by Vibra-Tech South. A total of five (5) shots were made with maximum recorded perticle velocity being 0.91 ips; acceleration being 1.48 g's with a perticle displacement of 0.002 inch. Maximum pounds of explosive was 7-1/2 lbs. with a maximum 1 lbs/delay = 1/2 lb. See Table 8 for tabulated blast data.
- 5.4-PH Vertical Rock Face: The vertical rock face between Service Bays 1 and 0 required rock removal to allow for the minimum required concrete thickness in the vertical wall. This wedge of rock was line drilled with hand-held air-powered rock drills and removed using Bristar.

TABLE 8

DT 100	ONTA	_	POWERHOUSE	EXCAVATION
RLAST	DATA	_	FUNDAMOUSE	TUCK LUTTON

Date	1-19-82 (#1)	1-19-82 (#2)	1-19-82 (#3)
Number Holes	28	37	24
Diameter (inches) Depth (inches) Spacing (feet) Burden (feet)	1-1/2 9"-14" 18" 24"	1-1/2 15"-18" 2' 2'	2-1/2 15"-18" 2' 2'
Explosives Weight ()	lbs)		
40% Gel	5-1/3	7	4-1/2
Total	5-1/3	7	4-1/2
Atlas Rockmaster De	lays 1,3-6,8-16	1,3-6,8-21	1,2-6,8-16
Maximum Holes/Delay	2	2	2
Maximum Lbs./Delay	6 02.	6 oz.	6 02.
Stemming (inches)	6"-10"	3"-15"	3"-15"
Theoretical Delay Duration of Blast	25 ms 550 ms	25 ms 750 ms	25 ms 550 ms

TABLE 8 (CONTINUED)

BLAST DATA - POWERHOUSE EXCAVATION

Date	1-20-82 (#4)	1-20-82 (#5)
Number Holes	20	12
Diameter (inches) Depth (inches) Spacing (feet) Burden (feet)	1-1/2 24"-9" 2 2	1-1/2 16" 2 2
Explosives Weight (lbs)		
40% Gel	7-1/2	2-1/3
Total	7-1/2	2-1/3
Atlas Rockmaster Delay	1-5,7-19,21,25	7-13
Maximum Holes/Dalay	1	2
Maximum Lbs./Delay	8 oz.	6 oz.
Stemming (inches)	1.5"-6"	13"
Theoretical Delay Duration of Blast	25 ms 975 ms	25 ms 250 ms

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6.0-CD FOUNDATION EXPLORATIONS DURING CONSTRUCTION - CONCRETE DAM

The excavation contract included an item for four-inch rock core "to determine the condition of the rock prior to and during excavation". In the fall 1977, 14 borings were drilled to explore unusual conditions revealed by clearing operations. These were limited primarily to near vertical fault and shear zones.

During early excavation on the South Carolina abutment, a number of "clay seems" had been encountered. Since some of these seems had been quite extensive, it was necessary to determine their existence and areal extent below planned foundation grade. Although they were difficult to pick up in core borings, a series of holes was laid out in both abutments and the spillway area, where deep horizontal joints would not be intersected by the excavation. Eventually, 74 borings were done during this program.

Four-inch borings done in the west abutment area encountered clean, leached joints which dipped at a low angle below design grade. To explore these joints extensively and economically, the excavation contract was modified to allow air-trac drilling without water. The drills were observed by Resident Office personnel and depths were recorded for color changes in cuttings, loss of cuttings, and water or mud production. The recorded depths and types of phenomena were then correlated and three-point problems worked out to define the attitude of the relief joints. The holes were drilled on 20-foot centers in Monoliths 1 and 4. There were approximately three joints striking across the dam axis and dipping in an easterly direction parallel to the axis. These joints were exposed when the 35-foot "step-up" face between Blocks 7 and 8 was revealed by excavation.

No exploratory investigations were done on the concrete dam contract.

6.0-EE FOUNDATION EXPLORATIONS DURING CONSTRUCTION - EARTH EMBANKMENTS

The initial contract estimate for exploratory borings called for 2,300 lineal feet of NX size coring. Field conditions indicated the need for additional coring and a modification for 3,100 lineal feet of additional drilling was issued. Of the total contract amount of 5,400 lineal feet of exploratory drilling, a total of 4,629.2 feet were actually drilled. The locations of some of the exploration drill holes are shown on Plates 67 and 68.

Logs of a limited number of the exploratory borings made by the Contractor are included as Appendix J. The contract specifically required the Contractor to provide complete drill logs for each boring made, and went into great detail regarding the educational and experience requirements for the Contractor Quality Control organization's resident Geologist, as well as listing all items which were to be included on the drill logs and on the core boxes. For reasons unknown these requirements were not met, and hence, much of the core recovered from the exploratory borings is of extremely limited value.

6.0-PH FOUNDATION EXPLORATIONS DURING CONSTRUCTION - POWERHOUSE

During construction of the Richard B. Russell powerhouse, foundation investigations consisted entirely of personal inspection and evaluation of the foundation by the Project Geologists. The entire foundation area was mapped geologically and personal inspection and approval by one of the Project Geologists was required prior to concrete placement. In addition, drilling operations for rock bolts, relief wells, and grout holes were monitored periodically and driller's reports, drill cuttings, hydraulic pressure test results, and grout operations were evaluated by the Project Geologists. No borings or tests were made for the express purpose of investigating foundation conditions during construction of the powerhouse.

7.0-CD UNUSUAL OR UNANTICIPATED CONDITIONS ENCOUNTERED DURING CONSTRUCTION - CONCRETE DAM

Several geologic conditions encountered in the concrete dam foundation caused problems in excavation and foundation preparation. However, these conditions were not totally unexpected. The extent or location of these features was not predicted in some cases, but their presence or the presence of similar features had been noted by SASEN-GG. These conditions will be covered in Section 9.0-CD.

One geological feature which seemed unusual to the resident staff was the presence of areas in the foundation which had been subjected to rotational stress, probably in a near horizontal plane. Evidence of this can be seen in the distorted metadiabase dikes and in tension cracks or tears within the metadacite country rock. Such evidence was found in Monoliths 13, 15, 16, 24, 25, 27, and 28, but it is most convincing in Monolith 21. There it appears that the semiplastic rock (probably during a metamorphic phase) was twisted to bend the previously intruded dikes. At sites of most extreme dislocation small tears, sometimes an echelon, were produced. Also found within the disturbed area were blebs of "bull" quartz and pockets of euhedral, pink calcite.

7.0-EE UNUSUAL OR UNANTICIPATED GEOLOGIC CONDITIONS ENCOUNTERED DURING CONSTRUCTION - EARTH EMBANGMENTS

Considering the sheer size of the Richard B. Russell embankments and the geologic terrane within which the project is located, suprisingly few unusual or unanticipated geologic features were encountered during construction. The specialized treatment of selected areas due to adverse geologic conditions is described in the following paragraphs by type of treatment.

7.1.EE <u>Diversion Channel and Fault Zone Treatment</u>: A major fault zone bisects the <u>Georgia embankment</u> in the vicinity of the old diversion channel. This fault trends approximately N10°E and dips about 50 degrees to the northwest. Movement along the fault appears to consist of both a right lateral strike component and a normal dip component. The fault itself has been intruded by a felsic dike shortly after faulting. Appendix A contains the complete petrographic and stress history analyses performed by the South Atlantic Division Laboratory.

Although the rock types on either side of the fault were identical (i.e., quartz-feldspar gneiss cut by mafic dikes), the appearance and geotechnical properties were different. The weathering profile was much more pronounced and extended to much greater depths in the hanging wall than in the footwall. The magnitude of this difference was clearly illustrated during excavation of the fault for dental treatment. Excavation was performed with a Cat 245 backhoe which experienced little or no difficulty in excavating the hanging wall; however, excavation of the footwall could be performed only with extreme difficulty, if at all.

In addition to the fault zone itself, the embankment crossing of the diversion channel was of great concern to the designers as well as field engineers. This was the highest section of the embankment and had abrupt

foundation elevation changes on either side. For these reasons, the most thorough, conservative treatment was performed in this area.

- 7.1.1-EE Excavation: The steep side slopes of the diversion channel were flattened to approximately IV on 2H by systematic drilling and blasting. The product was a mixture of sound rock and firm rock that was wasted, since the need for these materials had been met. (Contractor was permitted to process some for use as road plating material.) The result was a "stair-step" foundation on the east and west slopes. The bottom of the channel was excavated from elevation 330.0 to 315.0 approximately. The hanging wall of the fault was further excavated to elevation 309.0 after grouting had been completed. The fault proper was taken down to elevation 305.0 for a width of approximately 10 feet and this "trench" was then backfilled with concrete and the entire portion of the diversion channel underlying the impervious zone of the embankment received shotcrete (refer to Appendices D thru F).
- 7.1.2-EE <u>Drilling and Grouting</u>: Following cleanup, inspection, and approval, the area was released to a comprehensive drilling and grouting program. The curtain grouting consisted of a single line on the slopes and two lines across the bottom of the channel. Grouting was performed to a depth of 125 feet below the surface in 3 zones using stage-grouting. Significant surface leaks were noted in the area west of the fault in highly weathered rock, even though 3 to 5-foot nipples were used. Consolidation grouting was performed in the area downstream of centerline and west of the fault (Plate 109). Grouting was taken to a point below the fault, using stop-grouting. Two consolidation grout lines were added on the west side and one contract line deleted on the east side. Full details of consolidation grouting are covered later in this section.
- 7.1.3-EE Filters: Plans called for the downstream side of the cutoff trench in the bottom of the channel to receive filters and shotcrete. During the inspection on 5 May 1982, it was recommended that this configuration of filters be re-evaluated and consideration be given to shotcreting the downstream wall. SASEN-GS later directed that the downstream wall from elevation 310.0 to 330.0, approximately, be shotcreted and the filters begin at the 330.0 level.
- 7.1.4-EE Consolidation Grouting: The hanging wall of the Diversion Channel Fault downstream of the main grout curtain was consolidated by stop grouting. Plate 109 shows the grout hole layout profile illustrating the geometrical relationship between the consolidated zone and athe fault plane.

Stop grouting (as opposed to stage grouting) was the method chosen for this work as it was believed to be more effective for consolidation. Table 9 lists the grout takes and provides a statistical analysis of the grouting operation. Exploratory borings taken after grouting indicated that the grouting program was effective and that the desired end produce was obtained

7.2-EE Soft Pockets Near Diversion Channel: Around the centerline of the embankment between Stations 12+00 and 13+00 were several pockets of weathered rock. This rippable material was similar to, but stronger than that encountered in the bathtub area. It differed from the bathtub in that

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TABLE 9

CONSOLIDATION GROUTING

HOLE NO.	DEPTH(feet)	TOTAL TAKE(cu. ft.)	TAKE/LF(cu.	ft.) REMARKS
U-1	25.0	0.0	0.0	
U-2	25.0	0.0	0.0	
U-3	25.0	0.09	0.004	
V-1	25.0	4.18	0.167	
V-2	25.0	0.0	0.0	
V - 3	25.0	0.0	0.0	
W-1	45.0	6.30	0.140	Surface Leak
w-2	45.0	3.54	0.079	Comm w/V1, W3
W-3	45.0	0.36	0.008	
x-1 1/2	40.0	17.84	0.446	
1/1	65.0	5.12	0.205	Surface Leak
x-2 1/2	30.0	2.71	0.090	
1/1		0.28	0.008	
x-3 1/2	30.0	0.14	0.005	Surface Leak
1/1		3,42	0.098	
Y-1 1/2	45.0	0.36	0.008	Surface beak
1/1		0.72	0.018	
Y-2 1/2	45.0	7.36	0.164	Surface Leak
1/1		0.72	0.009	
Y-3 1/2	45.0	0.09	0.002	
1/1		0.00	0.0	Surface Leak
Z-1 1/3	40.0	19.33	0.483	Surrace Bear.
1/2	65.0	5.0	0.200	
1/1	125.0	2.67	0.045	
Z-2 1/3		0.27	0.007	
1/2	100.0	0.09	0.002	
1/1	1 125.0	0.0	0.0	
Z-3 1/3	3 40.0	0.18	0.005	
1/3	2 100.0	0.0	0.0	
1/	1 125.0	0.0	0.0	

TABLE 9 (CONTINUED)

CONSOLIDATION GROUTING

HOLE NO.	DEPTH(feet)	TOTAL TAKE(cu. ft.)	TAKE/LF(cu.	ft.) REMARKS
ZZ-1 1/3	60.0	0.45	0.008	
1/2	90.0	0.0	0.0	
1/1	125.0	2.45	0.070	
ZZ-2 1/1	125.0	5.18	0.041	Comm W/ZZ-3
ZZ~3 1/1	125.0	0.45	0.004	Surface Leak
V-1.5	25.0	0.09	0.004	
W-2.5	20.0	0.0	0.0	
	45.0	0.0	0.0	
X~1.5	40.0	0.18	0.005	
X-2.5 1/2	40.0	0.0	0.0	
1/1	65.0	0.18	0.007	
Y-1.5 1/3	50.0	0.18	0.004	
1/2	80.0	0.0	0.0	
1/1	105.0	0.18	0.007	
2-1.5 1/3	40.0	0.18	0.005	
1/2	80.0	0.09	0.002	
1/1	105.0	0.09	0.004	

TOTAL GROUT HOLES DRILLED:

1,915 LF

TOTAL GROUT TAKE:

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90.11 cu. ft.

MEAN GROUT TAKE:

0.047 (cu. ft./LF)

STANDARD DEVIATION:

0.103

it was not saturated with groundwater at the time it was grouted. Although core holes were drilled in and near these some that jointed areas, the excessive weathering could not be related to an unusual geologic feature. Three extra lines of 20-foot deep holes were grouted upstream from Stations 11+80 to 13+00 to assure seepage cutoff in this rock.

7.3-EE "Bathtub": An area of severe weathering of the foundation rock was encountered in the Georgia west section from approximately Station 5+50 to Station 6+50 (see Plate 73). This area is the convergence point of three dikes resulting in weathering far greater than the surrounding area. Station 5+50 was the end point of the transition trench slope; therefore areas unstation were specified to be taken to ripper refusal. Since ripper refusal was being approached in surrounding areas, the contractor was directed to continue in this area, although the "design grade" had been passed. Since there were rock knobs above "grade" being encountered routinely, the possibility of not being able to obtain rippable rock in this section was not determined until the excavation was near elevation 395, about 10 feet below "grade". ("Grade" is taken as that elevation shown on Plate EE-25 of the Contract Plans, an anticipated elevation of ripper refusal.) At that time, an inspection was made of the area by field, district, and diversion personnel. On 16 May 1980, following cleanup of the area, and obtaining several exploratory borings, the decision was made to stop excavation and begin treatment. Since the area was seeping a considerable amount of water, some method of controlling the water was necessary during fill placement. A system of pumps and a concrete cap was agreed upon and the area released to drilling and grouting. A triple grout line was directed to be used and the grout curtain was shifted upstream to coincide with one of the dikes. The shift and triple line ended near Station 7+00. Additionally, a single line to a depth of 25 feet was installed along the upstream edge of the "bathtub" to help control seepage (ineffective). Drilling and grouting were completed in this area in late November 1980.

Details for the special foundation treatment for the "Bathtub" area were contained in Modification H (PO17). In June 1983, the contractor began "mucking out" the accumulation of water and mud from the area. Since this was the lowest point in the section and located at the toe of the Georgia transition slope, a considerable amount of material had settled here. The tedious work was slowed by rain and additional wash-in of material. The base rock was very weathered and deteriorated quickly. Final preparation was parformed only hours prior to concrete placement. Treatment of the area was as follows:

- a. Final Cleanup and Wash-down: This resulted in an overall average foundation elevation of 385.0 feet.
- b. <u>Seepage Control</u>: A trench was dug along the upstream side and a 36-inch corrugated metal pipe installed vertically at each end (No. 1 and No. 2). Numerous 1-inch holes were burned in the pipe to permit water flow into the sump. A similar sump (No. 3) and trench were located on the downstream side of the area over a flowing seep. The trenches were then filled with 3-inch concrete aggregate. An additional sump (No. 4) was installed in a similar manner upstream of the "bathtub", not connected to the other sumps, needed for other seepage control.

- c. Concrete Pad: The rock drain, above, was then covered with plastic and the area covered with concrete (the trenches did not connect) with a thickness of approximately 3 feet. A total of 114 cubic yards of concrete was used. The top of the concrete pad was not level, but had an average elevation of about 389.0.
- d. <u>Sand Drain</u>: Water flowing out of the upstream slope remained to be controlled. After the concrete had cured, the plastic was removed from the top of the concrete aggregate and approximately 1 foot of aggregate removed. This was replaced with 3 to 6 inches of coarse filter. This was followed with filter sand. The sand was saturated and compacted beneath the top of the pad. A 12-inch wide sand drain was placed adjacent to the upstream slope to intercept seepage and carry it to the sumps. Impervious fill was placed on the concrete pad. The sand drain extended to elevation 395.0.
- e. <u>Fill Placement</u>: Following the treatment, fill placement continued until drainage out of the area was obtained, approximately 2 days work in August and November 1981. Work in the area then ceased until June 1982. Pumping of the sumps continued during this time since the fill level did not exceed the piezometric head.
- f. Capping: During fill placement in the summer of 1982, the piezometric level in the sumps was monitored by cutting the pumps off and allowing water to rise in the sumps. When the fill reached elevation 415.0, it was determined that the sumps were no longer needed and could be sealed. Grout pipes had be n installed in the trenches for grouting purposes. A 1:1 grout mix was pumped into the grout pipes (gravity, no added pressure) until grout showed in the sumps. This was allowed to take its initial set, then the sumps were filled with concrete (total 32 cubic yards).

g. Summary Data:

SUMP	BOTTOM OF TRENCH	TOP OF CONCRETE	TOP OF SAND DRAIN	TOP OF FINAL RISER	BAGS OF CEMENT GROUT
1	383.3	388.6	395.0	417.5	57
2	383.0	388.2	395.0	418.0	82
3	384.2	390.2	N/A	418.1	102
4	390.5	N/A	395.0	418.1	114

7.4-EE Relief Joints: There were several clean, unfilled relief joints on the Georgia abutment under both the concrete dam and the embankment. These sheet joints separated otherwise sound foundation rock into several layers like an onion. When the Georgia east portion of the Georgia embankment was excavated, the contractor's bulldozer ripped along these sheet joints for some distance. When it was noted that two of the joints had been chased to some depth, it was decided to end ripping and excavation at a cross-dike near the 22+00 station of the dam axis. At this point, there was a 10-12-foot stepup on a steep (70-80 degree) slope. After the area was grouted, an unformed fillet was placed to about a foot above the upper joint, furnishing a low angle face against which fill could be easily

compacted. Grout communication was noted in the first zone between holes 200 feet apart below this face, probably indicating another relief joint at a shallow depth.

7.5-EE Formed Concrete Fillst - South Carolina: Approximately 7,500 cubic yerds of rock excavation was required in South Carolina to remove a rock knob in the cutoff trench. The rock on the upstream well was characteristically blocky, jointed, loose, and had numerous clay seams. The blasting left a very rough blocky, upstream slope having large, high overhangs and open seams throughout the blast zone. Due to the geology of the area and the presence of clay seams, it was felt that the jointed nature of the rock was not caused by blasting, although some of the seams were probably opened further. Treatment considerations included additional excavation in the slope, pulling loose blocks out of the slope, and placing a formed fillet to cover the area. Since any excavation attempt may have accentuated the problem, the decision was made to use a formed concrete fillet, approximately 15 feet high. The fillet was placed in two lifts, with dowels, and had a total of 244 cubic yards of concrete. This was accomplished under Modification PO12 (L).

7.0-PH SPECIAL OR UNUSUAL CONDITIONS DURING CONSTRUCTION - POWERHOUSE

The foundation of the Russell powerhouse was found to be rather straight forward from a geologic point of view and no unusual nor unexpected conditions were encountered during construction.

8.0-CD FOUNDATION ANCHORS AND ROCK BOLTS - CONCRETE DAM

Rock anchors were installed on both the excavation and concrete dam contracts. The anchors on the excavation contract were for prevention of slides as described above. Foundation anchors and well anchors were installed on the concrete dam contract.

The rock anchors installed under the excavation contract were Dywidag threadbars of 1-inch diameter and 60,000 psi yield strength. They were installed in 1-5/8-inch diameter holes with two 12-inch long Celtite fast set, two-component, polyester resin cartridges in the bottom of the hole. After the resin was allowed to set, the anchors were tensioned with a torque wrench to 2/3 of the bar yield strength (21,500 lf.). Tension was locked into the bolt by a 1-7/16-inch haxagonal nut against a 6" x 6" steel plate 3/8 inch thick. Corrosion protection for the stressing length was not provided.

Foundation anchors were installed in only one of the thirty-two monoliths of the concrete dam. Block 26, unlike the others, was divided into two parts by a construction joint. The downstream part of the block formed the flip bucket for the easternmost tainter gate. Since the flow of water over the flip bucket would impart an overturning moment to this monolith, it was considered necessary to anchor the concrete firmly to the foundation rock. This was done by grouting number 11 reinforcing steel bars into 10-foot deep holes in the downstream area of Block 26. The projecting ends of the bars were deformed into a J-shape prior to installation and the concrete was placed around them after the grout cured. The broom grout mix that was utilized to treat the foundation prior to concrete placement on rock was used to cement the bars into the holes.

The rest of the anchors installed on the concrete dam contract were used to secure the tailrace and erection bay hang-on walls. Two types of anchor were used. Twenty-seven one-inch diameter billet steel bars were used as untensioned dowels in the lower part of the tailrace wall. These dowels were installed 3 feet into rock using Celtite fast set resin cartridges prior to placement of the concrete wall. Hooked attachements project into the wall to about 6° from the outer surface of the wall.

On the upper part of the tailrace wall and on the erection bay wall, post-tensioned anchors were used. There were 145 of these in the erection bay and 117 in the tailrace wall. The tailrace wall anchors were originally designed to be installed before placement of the well, but proved to be difficult to install in the very irregular rock face. The contractor proposed, and the Corps accepted, a no-cost change to modify the contract to allow forming of blockouts in the well and installation of the anchors in these blockouts after the wall was in place. Thus, the same installation procedure was used on both walls.

The anchors were 1-1/4-inch Dywideg threadbars of 187,500 lb. ultimate strength. Lengths varying from 24 feet (20' in rock) to 35 feet (31' in rock) were installed in holes of 2-inch dismeter. A combination of fast and slow set Celtite cartridges was inserted into the hole to give full encapsulation with a 12-foot (minimum) anchor zone. After insertion of the thread-ber and setting of the anchor zone resin, the bar was stressed to

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150,000 lbs. and locked off at 112,000 lbs. by a hazagonal nut against a 9' \times 9' \times 1-3/4 anchor plate. When all the anchors had been installed in a wall, the contractor flame-cut the anchor 3"-6" inside the blockout and filled the blockouts with concrete. Flame-cutting was allowed only after the contractor proved that he could cut the threadbers with a cutting torch with less residual heat than that involved in sawing them with an abrasive disc. Plate 128 contains installation details of rock anchors used in the tailrace and erection bay hang-on walls.

8.0-PH FOUNDATION ANCHORS AND ROCK BOLTS - POWERHOUSE

- 8.1-PH General: A relatively large number of rock support elements were employed in the powerhouse structure so that structural stability criteria were met. Both deformed bar rock bolts and multi-strand, high-capacity rock anchors were utilized in meeting these criteria.
- 8.2-PH Deformed Bar Rock Bolts: Deformed bar rock bolts (also referred to as "thread bar rock bolts") constitute, by far, the majority of the rock support elements installed in the powerhouse. These bolts are 1 1/4 inches in diameter and are various lengths, grade 150, and utilized a polyester resin grout. Each rock bolt assembly consists of the bolt, a steel bearing plate, a hexagonal nut, a sufficient number of fast set resin cartridges for the anchor zone, and sufficient number of slow set resin cartridges to fill the remainder of the drill hole. A typical installation sequence is given below:
 - 1. Drill and clean hole (percussion drilling only was allowed).
- Insert the required number of fast set and slow set polyester resin grout cartridges.
- Insert the bolt into the hole, using the air-track drill to apply both rotation and downward pressure.
- 4. Once the bolt is inserted to the required depth, continue rotation until the anchor zone grout sets up.
- 5. Remove the drill, install the hydraulic stressing jack and stress the anchor with the required amount of force.
- Tighten the hexagonal lock-off nut and remove the stressing jack.

Plates 127 and 128 show the locations of the deformed bar rock bolts and Plate 129 shows typical details.

- 8.2.1-PH <u>Draft Tube Slab and Service Bay Slab Rock Bolts</u>: Plate 127 shows the locations for these bolts. These areas utilized 35-foot long bolts except in selected areas in the vicinity of the unwatering drainage trench where longer bolts were required so that the 20-foot anchor zone in rock criteria would be met.
- 8.2.2-PH Design Requirement for Anchoring the Draft Tube and Service
 Bey Slebs: Each generator bay monolith was designed as a "stand alone"

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gravity structure with the dead load of each monolith being of such magnitude that all anticipated hydrostatic uplift pressures would be compensated for and hence, no external anchorage would be required. However, the draft tube and service bay slabs are not structurally integrated with the monoliths and do not transmit foundation loads, contribute weight, nor resist uplift forces. Therefore, a system of foundation drains (described in Section 10.0-PH and shown on Plate 126 and rock bolts (Plate 127) was used to overcome hydrostatic uplift forces within these areas. The following design assumptions were made in the layout of the rock bolts for these areas:

- a. Foundation drains were assumed to be 50% effective.
- b. The maximum expectable tailwater elevation was assumed to be Clarks Hill standard project flood (elev. 342.0).
- 8.2.3-PH Erection Bay Rock Bolts: Deformed bar rock bolts were used in both the erection bay slab (vertical, 35-foot lengths) and the erection bay substructure wall (horizontal, variable lengths) to compensate for hydrostatic pressures acting against these structural elements. Plates 127 and 128 depict the spatial arrangement of bolts in these portions of the structure. The erection bay substructure wall and tailrace hang-on wall was built under the concrete dam contract by the concrete dam contractor (Dravo-Groves). Initial plans for the substructure wall and tailrace hang-on wall called for the installation of rock bolts directly against the rock face. The contractor submitted a proposal to install the bolts through the concrete wall; this proposal was evaluated and accepted by the Corps and was incorporated into Modification FF. Plate 128 depicts the locations and details of these bolts.
- 8.2.4-PH Spillway Training Wall: Rock bolting and anchoring for the spillway training wall is covered in detail in Section 4.0-PH.
- 8.3-PH Multi-Strand, High-Capacity Rock Anchors: Rock anchors are used in the powerhouse in Units 5 through 8 and in the spillway training wall. Thirty strand anchors are located in the units while twenty-three strand anchors are to be found in the spillway training wall. Each strand is 1/2" in diameter and is composed of 7 wires conforming to ASIM Ad16 for prestressing steel. Each strand possesses a minimum ultimate strength of 41.3 kips. Installation procedures for both types of anchors are similar with only the ultimate and lock-off applied forces differing. Plate 129 depicts typical as-installed details while Plate 127 depicts the locations.
- 8.3.1-PH Installation Procedure: The following is the installation sequence for all of the stranded rock anchors employed in the powerhouse:
 - 1. Drill holes with down-hole hammer.

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- 2. Pressure test holes, grout, and re-drill if necessary.
- 3. Insert anchor and perform first-stage grouting.
- 4. Test grout cylinders for strength (>3000 psi).

- 5. Stress anchors to 80% of ultimate strength hold for 10 minutes and lock-off at 70% of ultimate.
- 6. After 7-day waiting period, perform lift-off test. If lift-off test reveals no relaxation, got to step 7. If lift-off indicates that relaxation has taken place, go back to step 5.
 - 7. Perform second-stage grouting.
- 8.3.2-PH Grout Materials: The use of a high-early strength, non-shrink, Portland cement-based grout was specified for the stranded rock anchors. The contract specified a site-mixed grout using a number of admixtures, but the contractor proposed the use of a pre-mixed, commerically available grout. After evaluation of the manufacturer's literature and limited testing at both the Project Office and the SAD Lab, approval was given to the contractor to use the commercially available grout. This approval was contingent upon satisfactory performance and at no additional cost to the Government. During the installation of the anchors using this grout, no problems were encountered. It is felt that performance probably exceeded what would have been attained had the original grout mix been used.
- 8.3.3-PH Engineering Considerations for Use of Stranded Rock Anchors in Units 5 Through 8: As stated previously each powerhouse monolith was designed as a stand-alone gravity structure as the total dead load of each unit was sufficient to prevent hydrostatic uplift. However, since Units 5-8 will consist only of the skeletonized structure for a number of years, it was imperative that the reduced dead load of these units be compensated for so as to prevent uplift. A number of different schemes were analyzed for sufficiency, constructability, maintainability, and cost effectiveness. On all counts, the use of rock anchors proved to be the most viable alternative. Although the time span in which the anchors are required to function is relatively short (less than 10 years), they are a permanent part of the structure and should function properly for the entire life of the project.
- 8.3.4-PH Stranded Rock Anchors in the Spillway Training Wall: Rock anchorage requirements and considerations for the spillway training wall are covered in detail in Section 4.0-PH.

9.0-CD CHARACTER OF FOUNDATION - CONCRETE DAM

9.1-CD <u>Foundation Surface</u>: After foundation preparation, most of the Richard B. Russell foundation surface was very irregular. Exceptions to this were the parts of Monoliths 2 and 3 which were excavated with a backhoe along a low engle joint, and Monolith 26, excavated along a clay seam. These areas were fairly smooth after drummy plates and slabs of rock had been removed.

The extensive jointing, faulting, and shear zones in the dam area, along with the multiple dikes and dike swarms gave the rock a tendency to break in an irregular, blocky manner. This was intensified by the release of high velocity gasses from blasting agents and explosives used in modern construction practice. The combination of hard, strong rock with pre-existing, weaker planes made it difficult to achieve regular excavation surfaces, even when presplitting techniques were used.

Design slopes flatter than 1V on 1H usually broke below grade in a blocky or stairstepped menner. This was probably due to the presence of incipient near-horizontal planes of breakage in the foundation rock. These planes may be created by stress relief or tectonic forces, but they are present in most rock masses in the area, including the Elberton Granite. When explosives were detonated, these planes furnished an easy exit for expanding gasses. Holes downslope propagated breaks well into and through the plane of the design slope. When final rock removal had been accomplished to remove rock which had been loosened, the effect was even more pronounced. Areas where the stairstep effect was most apparent were the 1V on 3.5H slope downstream of Monoliths 8 through 16, the 1V on 3.18H slope downstream of Monoliths 6 and 7, and the sump in Monolith 9 (excavated by Dravo-Groves).

Where the design surface was horizontal, the final surface was also uneven. Preferential breakage along the natural discontinuities described above resulted in a foundation with many abrupt changes in slope. The felsic rocks were more resistant to breakage than were the mafic dikes. Percussion tools such as horrams and pavement breakers were used as little as possible in the brittle metadiabase. Another factor which caused an irregular surface was overdepth blast holes. Although the contractor maintained good drilling control on most blasts to grade, some holes were drilled below specified tolerances. During loading, holes determined to be overdepth were bottom-stemmed back to grade and loaded as planned. However, a few of these holes were not identified. Shooting an overdepth hole resulted in a depression, usually conical, in the foundation.

Foundation surfaces IV on IH or steeper were commonly more irregular than what is usual for presplit slopes. The rock in these faces broke along joints at varying attitudes. The presplit face between Monoliths 7 and 8 was the only one between monoliths which was excavated close to design. The faces upstream of Monoliths 8 through 13 also broke out well. Some of the smaller vertical faces between monoliths broke irregularly along their entire length. It was necessary to make "plug pours" in several of the monoliths on the east abutment to replace rock broken off vertical faces between monoliths.

- 9.2-CD <u>Condition of Foundation Rock</u>: The Richard B. Russell concrete dam was founded entirely on rock. With the exception of a few shear zones which were hydrothermally altered or filled with softer minerals, the rock was sound. The engineering characteristics were within those reported in the Design Mamorandum 8 Geology and summarized in Section 3.5 of this report.
- 9.3-CD Water: The measures described in Section 5.0-CD of this report were adequate to allow excavation of the dam and powerhouse in the dry. Some water entered the excavation from upstream of the intake and spillway blocks, but the quantity of flow was too low to cause problems. During concrete placement, water problems were encountered in the intake and spillway sections west of Block 24. Blocks 1-7 and 24-32 were founded at higher elevations. Seepage and natural runoff were intercepted and ponded by a low dike/road 20 to 30 feet upstream of Blocks 10-23. The water level here was probably at or above 300 MSL.

In Blocks 8-23, where seepage along joints or seams was very slight, concrete was marely placed and vibrated well against the seam. In these cases, the weight and strength of the concrete was judged to be sufficient to stop these minor leaks without special treatment. Blocks in which no special treatment was needed were Block 8 (where wet weather seeps were present along the 7/8 face), Block 10 (a few seeps were present on the upstream face), Block 11, Block 17, Block 20, and Block 21.

Where seepage was present along the upstream face at fairly low pressures, visqueen sheeting was used to separate the water seeps from the concrete. This was done by placing the visqueen on the face immediately before the last buckets of concrete were placed in the block. (Placement proceeded from downstream to upstream). Then the last concrete was placed against the visqueen, causing the water to rise at the upstream face. This was done in Blocks 9, 14, and 18.

Upstream of Blocks 13, 14, and 15, a small sump was created by a backhoe to collect water. The sump was pumped nearly dry during placement. This eliminated the necessity for special treatment in Blocks 13-15.

Where moderate flows were present on the upstream face, french drains were made by pinning 6" wide steel strips vertically to the foundation approximately 1 foot from the face. These were filled with uniformly graded 3" aggregate and a visqueen sheet was placed over them. Pipes, or 55-gallon drums, were used to remove or pump water from these drains. At least 1 supply and 1 return pipe were always installed.

After one or more lifts of concrete were in place and cured, the drains were pressure grouted with 1:1 grout until grout was seen in the return lines. Blocks 12 and 22 were treated this way until grout was seen in the return lines. Blocks 19 and 16 had spot flows. Block 16 was the only monolith with a leak in the horizontal foundation. This leak was west and downstream of the inverted plumb bob, below the present inspection gallery. These spot flow areas were surrounded or filled with clean 3" gravel with 55-gallon drums, used to blockout a space for grout pipes. They were treated as the french drain above, but after being confined by one or more lifts, they were first grouted, then concreted up with a 3/4" mix.

Locations of Treatment Described:

Block 12 - Against the eastern 2/3 of the upstream face, relief pipe at the corner near the 12/13 joint.

Block 16 - Immediately below the inspection gallery as the gallery turns a corner to enter Block 17, near the inverted plumb bob.

Block 19 - Against the upstream face about 10 feet from the 18/19 joint.

Block 22 - Against the upstream face in the western corner.

9.4-CD <u>Special or Unusual Conditions</u>: Foundation conditions which required special consideration and/or treatment in the construction of the Russell Dam include seams, unexpected areas of deteriorated rock, and water problems.

The clean joints in the Georgia abutment have been mentioned in the Foundation Exploration and Geology sections of this report. They were discovered by examining 4-inch core taken under the excavation contract. The joints were low angle, irregular, and leached. The air-trac exploration described in the Exploration Section was initiated to determine the extent and attitude of these joints. It was discovered that there were (probably) three seams, surfacing in Monoliths 5 or 6, Monoliths 2 and 3, and outside the dam under the Georgia embankment, and exiting the stepup face between Blocks 7 and 8. The most probably dip on these subparallel joints was toward the river and slightly upstream. Because they were clean, irregular (with good rock to rock contact), and dipped upstream, it was decided that these joints posed no problem to the integrity of the dam, except that if ungrouted, they would allow water to seep through the foundation. They were penetrated by curtain grout holes, and based on communication in the first zone, were grouted up. No further treatment was considered necessary.

An unexpected area of deteriorated rock was discovered during excavation of Monoliths 2 and 3. This rock, located between one of the joints mentioned above and a shear zone, showed evidence of hydrothermal alteration and severe weathering. Possibly, the joint allowed easy penetration by hydrothermal solutions from the (also hydrothermally altered) shear zone which caused decomposition of the rock and allowed (later) rapid weathering. The rock below the joint was hard, fresh, and only slightly weathered, so treatment beyond removal of the decomposed material and replacement with concrete was considered unnecessary.

A (probable) relief joint intersecting the upstream face of Blocks 13, 14, and 15 allowed pended water upstream to infiltrate into the excavation, depositing rusty-appearing iron salts on the foundation and causing minor water problems during construction of these blocks. The water was handled during placement as described in Section 10.0-CD of this report. The iron salts were cleaned from the foundation by sand-blasting and/or water-blasting at maximum force with great difficulty.

Another area of deteriorated rock which had to be removed and replaced was in the upstream part of Monolith 16 near the inverted plumb bob located in that block. The hydrothermally altered and weathered rock was above a

low angle shear zone. The area was excavated to sound rock. The geology and topography of that zone can be seen on the maps of the Block 16 area.

A foundation condition which caused considerable construction change was the presence of seams filled with shattered and decomposed rock in the spillway area. These seams were noted while logging core drilled on the excavation contract. Excavation grades were changed to remove rock above some of these seams, but some were left after the excavation contract was completed. Except where they occurred under trestle footers, the rock above them was removed under the concrete dam contract as part of the rough cleanup phase of foundation preparation. Where trestle footers were underlain by seams, the seam and the rock above it were removed to within a few feet of the footer. A small amount of rock was left to avoid undermining the footer. In Block 17, a 20' x 10' slab of rock was left under the northeast trestle footer. In Block 19, rock was removed to within 1 to 5 feet of the edge of the northeast footer, then the footer was surrounded by a concrete collar to protect it until the first lift of concrete could be placed.

During foundation rock removal in Monolith 22, a seam filled with mud and rock fragments was discovered. This seam was chased into Monolith 23, where it was tighter and contained no mud. Previous borings indicated that the seam stopped at a dike which crossed the monolith diagonally a short distance from the existing face. It was decided to discontinue rock removal at the point where there was no mud and there were fewer rock fragments, and treat the seam by grouting. This treatment is described in Section 10.1-CD.

Clay filled breaks in foundation rock also necessitated changes in construction. These "clay seams" were joints, relief joints, and irregular breaks at all orientations which were filled with a plastic, red clay. Analysis of the clay showed that it was primarily kaolinite, probably a product of chemical weathering of feldspars. It had probably been transported to the "seams" by the movement of groundwater, and then had settled out of suspension in all available open spaces. The "clay seams" were present in both abutments, but were a problem only in the South Carolina abutment where foundation grade had been set higher in the zone of weathering. In this abutment, the extensive, near-horizontal relief joints were the major problem. Grades were lowered to eliminate the seams and rock was removed under modification to found below them. Blocks 28, 27, and 26 were those overexcavated. Additional changes which were made to adapt the structure included additional grout holes, a change in the geometry of the inspection gallery, additional instrumentation, and a more extensive system of waterstops on the monolith construction joint between Blocks 28 and 29.

A final problem which caused minor difficulty was the presence of near vertical round-shouldered rifts filled with cobbles, sand, and gravel in the riverbed. These rifts were probably joints which had been widened by the action of the river's current and bedload on the rock adjacent to the existing open joints. The loose material in the rifts tended to fill in blast holes as they were drilled, causing the excavation contractor's blast holes to collapse as they were drilled.

Prior to any concrete placement for dam construction, a written record of final foundation approval was made after inspection of the foundation in

the field. Placement cards were kept for each monolith; however, these were not available for inclusion in this report.

9.0-EE CHARACTER OF THE FOUNDATION - EMBANKMENTS

- 9.1-EE <u>Foundation Surface</u>: The foundation surfaces upon which fill was placed can be categorized as one of the following: (1) in-situ residual soil in the transitions; (2) shotcrete or dental concrete treated firm rock and fractured sound rock; or, (3) unfractured sound rock. For details on the treatment of the various areas see Section 10.0-EE of this report.
- 9.2-EE <u>Engineering Characteristics of Foundation Materials</u>: Section 3.5 of this report summarizes the engineering characteristics of the foundation soil and rock for the embankments. Design Memorandum 8 Geology and Design Memorandum 17 Earth Embankments contain detailed studies and test results for the foundation and embankment materials for this project.
- 9.3-EE Groundwater Conditions Affecting Construction: Groundwater conditions affecting construction of the embankments were only of local extent, and with the exception of the "Bathtub", presented no major problems. Section 7.0-EE contains details on the "Bathtub" area and Section 5.0-EE contains details on dewatering in general. Also, refer to the Embankment Performance and Criteria Report for a more exhaustive treatment of dewatering of the formdation.

9.0-Pi: CHARACTER OF FOUNDATION - POWERHOUSE

- 9.1-PH Foundation Surface: The surface of the powerhouse foundation is very irregular due to the intense jointing of the rock. Joints within the foundation rock assume two major geometric styles which are:
- a. High-Angle, nearly vertical joints with two predominant strike directions. These joints have spacing that ranges from about 2 inches up to about 4 feet with an avarage spacing of about 1 foot.
- b. Subhorizontal Relief or Exfoliation Joints. These joints are subparallel to the topographic surface and were probably caused by removal of the overlying confining material by erosion and/or excavation. The spacing of these joints, many of which were only incipient until subjected to excavation equipment, ranges from about 2 inches to about 4 feet with an average spacing of about 1-1/2 feet.

As may be interpreted from the preceding paragraph, the powerhouse foundation consists of a "stair step" arrangement of joint blocks in those areas where the foundation contains a slope, and is more or less flat lying in the remainder. The "stair step" arrangement is particularly noticeable in the draft tube area of the powerhouse where the foundation slopes upstream and rather than being a detrimental characteristic, is in fact positive in that it increases the sliding resistance of the structure.

9.2-PH <u>Engineering Characteristics</u>: The powerhouse is founded on fresh unweathered rock. The rock is a hard, dense metamorphic rock whose strength and other engineering characteristics are more than adequate for the

structure being built upon it. For a more complete description of testing performed, refer to Section 3.5 and Table 3, page 28.

- 9.2.1-PH Additional Physical Tests: In addition to the rock tests mentioned in Section 3.5, tests for suitability of the rocks from the damsite for concrete aggregate were also performed. Although performed for another purpose, an analysis of the test results leads one to conclude that no adverse chemical reactions between the rock foundation and the concrete of the structure should take place. Further discussion of these tests are presented in detail in Design Memorandum 12 Construction Materials.
- 9.3-PH Groundwater: The occurrence of groundwater at a given site is a function of many interrelated factors, but perhaps the most important parameter is the permeability of the geologic formations present at the site. The Russell powerhouse is founded entirely in crystalline rocks whose primary permeability is very low and whose secondary permeability is also low due to the fact that most of the discontinuities occurring in the rocks are either tight, healed, filled with impermeable clays, or are discontinuous and not interconnected. Therefore, groundwater flows did not cause any great problems during construction of the powerhouse. Those groundwater inflows which were present were relatively small, localized affairs, which were controlled by french drains, described in Section 10.0-PH of this report. Foundation relief wells and weep holes are also described in Section 10.0-PH.
- 9.4-PH Special or Unusual Conditions: The foundation of the Richard B. Russell powerhouse was found to be rather straight forward from a geologic point of view and no unusual nor unexpected conditions were encountered during construction.

10.0-CD FOUNDATION TREATMENT - CONCRETE DAM

10.1-CD Grouting Prior to Concrete Placement: A 1" to 4" thick seam filled with mid, decomposed rock, and rock fragments was exposed in Block 22 and during rough cleanup in that block was chased into Block 23. The intended method of treatment was to remove the seam and rock above it. However, as this excavation proceeded deeper into Monolith 23, the seam proved to be tighter. Fresh rock chips, such as are produced by blasting, were the only filling. Boring logs indicated that this seam probably did not extend past a metadiabase dike which crossed the block diagonally about 25 feet from the existing face. Due to the change in thickness and character of the seam, it was decided to leave the 4 feet of sound rock above the seam in the foundation and grout the remaining open portion.

Modification P023 was issued to provide for this operation. The contractor was directed to:

- 1. Drill fifteen 3" diameter holes through seams.
- 2. Wash holes and intersected seam until wash water return is clean.
- 3. Grout 1-1/2" diameter pipes into the holes.
- 4. Construct french drains at the excavation face. Install 2-1/2" diameter return pipes.
 - 5. Fabricate and install "Box Type" uplift cell in lieu of cell #24.
 - 6. Place plug pour over box cell and french drain.
- 7. Clean surface of affected area to allow observation of joints and fissures.
- 8. Using a 1:1 mix, begin grouting pipes as directed. If grout return from any hole is noted, cap off that hole. When maximum pressure of 15 psi is reached and held, cap off that hole and connect to next hole as directed. Continue until return or refusal is obtained in all holes and 2-1/2" return pipes.
- 9. Four (4) hours after grouting, wash out uplift cell holes to full depth.
- 10.2-CD <u>Curtain Grouting</u>: The concrete dam grout curtain is a single line, three zone curtain with holes spaced 5 feet apart. A few quaternary holes were drilled 2.5 feet away from existing holes where communication or high takes warranted. These holes were drilled through 3" sleeves (positioned prior to concrete placement) angled 15 degrees from vertical in an upstream direction perpendicular to the dam axis. The uniformity of the curtain as described was interrupted in three places. At both ends of the dam, grout holes were fanned out to provide for overlap with the curtain grout holes under the earth embankments. These holes were drilled at variable angles in a vertical plane along the dam axis. Spacing on these holes was two to four feet. They are depicted on Plates 51 and 52. The grout curtain i, "tonolith 28 was also irregular. This monolith was lowered

drastically during excavation, while Monolith 29 was left at original design grade. The grout curtain in this area would have had a large "window" where the collar elevation of the grout holes stepped up over 40 feet if the original design were followed. The inspection gallery was redesigned to include a "T" shaped appendix, with the crosspiece of the T running parallel to the 28-29 stepup face. Horizontal, 20 degree, and 45 degree holes were used to overlap the curtain holes originating in Block 29. Three holes were added on the stair from the original inspection gallery to Monolith 29. This is depicted on Plate 52.

Construction of the curtain was done from a working (inspection) gallery approximately 5' above the design rock foundation. The sequence of work was to drill and grout holes on 20-foot centers in the first zone (40' deep), then to split with secondary holes 10' away in the same zone, then to split with tertiaries 5 feet away. The second zone (80' deep) was completed in a similar manner with primary and secondary holes, and the third zone (120' deep) was completed with primary holes only. These holes were further split on either side when the hole communicated with others over a considerable distance or where the hole took ten bags or more of grout in that zone.

Grout holes were drilled with air-driven CP-65 drills using diamond-set coring bits, 1.5 inches in diameter, with water as the drilling medium. The metadacite and quartz porphyry were very hard, drilling more slowly than western (U.S.) quartz monzonites and granites according to drillers who had worked in these rocks. Rates for the first 5 months of grout hole drilling averaged 12 feet of redrill (through the cement-bentonite mix described below) and 36.2 feet of rock in an 8-hour drill shift. The original face-set bits used often lasted only 4-10 feet, so the contractor went to an impregnated coring bit to prolong bit life. This did not, however, improve drilling rates much.

After drilling, the holes were washed and pressure tested. When a hole had been drilled to full depth it was washed for a short time through the drill string to get the cuttings out of the hole. Following this, the holes were pressure-tested with water for 5 minutes, using the maximum allowable pressure. If the hole communicated with another, it was washed until the exhaust water was clear.

When holes within a specific reach of curtain had been drilled, washed, and pressure tested, grouting operations began. Bentonite and water were mixed in a paddle-type mixer located outside the dam, then routed through a Galigher centrifugal pump to thoroughly mix and hydrate the bentonite. Comment was then added to the water-bentonite mix in the same mixer. From the mixer, the grout was dumped into a large (30+ cu. yd.) sump. A constantly-circulating grout line (driven by a Moyno pump) made up of high-pressure hose and 1-inch pipe then took the grout from the plant the sump. The grout was screened prior ering the sump.

Grout mixes were composed of varying amounts of portland cement and water with bentonite. Water/cement ratios varied from 5 cubic feet of water: 1 bag (94 lb.) of cement to 1:1, with the average mix being placed in approximately 3.5:1. The contract provided for bentonite to be added at a rate of 4% (by weight) of the cement in the mix. However, experimentation

showed that 2% bentonite gave a better strength with nearly the same low shrinkage ratio as the 4% mix. The 2% bentonite mix with portland cement and water was used throughout the job.

Grout was carried from the circulating line to the hole by means of a T-shaped header made up of iron pipe, valves, gauges, and high pressure hose. The crossbar of the T was hooked into the circulating line with a valve to control the exhaust side of the T and to maintain pressure in the supply line at as high a level as required to grout. The lower part of the T was hooked into the hole by means of a high pressure hose and a packer. Grout from the supply line want through a cutoff valve and a gauge (equipped with a gauge-saver) before entering the hose and packer. Pressure in the hole was maintained by the valve and monitored by the gauge during operations. Mechanical and pneumatic packers were used to connect the grout line to the hole. Mechanical packers were used in all holes and all zones, while pneumatic packers were used in the first zone of hole 8+27 and all noles with higher numbers (east from there to the South Carolina abutment) for reasons stated below.

After the extensive leaks described below in this section were encountered, it was decided to grout the rest of the holes (holes from Station 8+85 to the South Carolina end of the gallery) in two steps. A pneumatic packer would be set just below the concrete/rock interface and the hole grouted at maximum pressure starting with a thin 5:1 mix. After the hole refused (took 1 foot or less liquid in 10 minutes), the pneumatic packer was withdrawn and a mechanical packer fixed in the grout nipple at the collar. The hole was then grouted a second time starting with a 3:1 mix. In this way, it was hoped we could get the curtain grouting done in the normal way, and then grout any voids in the concrete with a thicker mix since consolidation grouting was what we desired here.

Pressures used during grouting were held to a maximum total pressure of 40 psi in the first zone, 80 psi in the second, and 120 psi in the third. Gauge pressure was calculated as maximum total pressure minus the weight of the full column of the grout being used at the time. The pressure at the header gauge was thus varied with the mix being used at the time. For example, in the third zone, a mix of 5:1 grout would be pressured an additional 58 PSI, while 1:1 grout in the same zone would be pressured an additional 33 psi. When grouting in the first zone was done within 100 feet of any uplift cell, that well was pressurized to 15 psi. In most cases, this seemed to prevent grout infiltration into the uplift cells. Where grout was seen in uplift cell lines, the line, and hopefully the cell, was washed out with water immediately after grouting.

Pertinent data on grouting operations beneath the concrete dam is summarized on Plates 51 and 52. Hole numbers used on the charts and in the text of this report were assigned on the basis of stationing along the dam axis (Station 0+00 is against the gallery wall in Monolith 1). The true location of a grout hole is usually within .5 feet along station of this index in the upstream gutter. However, where the working gallery runs perpendicular to the axis (making all holes essentially at the same station) the holes are numbered consecutively until one hole corresponds with its station. Then stations and numbers are again essentially identical.

The second secon

During curtain grouting operations, we had several problems with leaks. We noted communication of grout holes with construction joints (the joint between Blocks 2 and 3 and the joint between Monoliths 4 and 5), drain hole nipples prior to drilling of the drains (several places throughout the dam), porous areas in the gallery floor (in Blocks 1, 26, and 27), and outside the dam.

Three of these leaks may be of importance later. During pressure testing, hole 0+10 in the Georgia abutment fan communicated with the area outside the dam where fill was being placed. Fill placement was discontinued to allow excavation to and investigation of the leak. Concrete was removed for a few inches into the leak to reveal a pipe which was carrying water into the fill. It is probable that the grout hole, some 26 feet from the end of the dam was connected to the outside through a lift line (possibly through "honeycombed" concrete or a rock pocket) to a bracing pipe which had held up a form during concrete placement. Water under pressure had infiltrated through this connection, broken through an inch or less of concrete over the pipe opening, and penetrated the fill, collecting on the surface. The leak was repaired by fully exposing the pipe, attaching a valve thereto, and grouting the connection with 1:1 grout.

Hole 16+93 near the Monolith 28/29 construction joint also showed wide communication. When zone 1 of this hole was drilled, the driller reported a 4"-6" tool drop. Pressure testing of the hole revealed communication with another grout hole and an uplift cell in Block 29. Later testing with dyed water showed that the hole also communicated with a hole in the "crosspiece" of the T-shaped gallery in Monolith 28 and with a drain which ran along the 28/29 stepup face. This hole was grouted with 1:1 grout pigmented with iron oxide powder. An exploratory hole drilled later intersected 4" of the brick-red grout in a natural seam. This hole was drilled in the area where the probable relief joint in the South Carolina abutment terminated. The hole probably intersected the joint, affording connection with another hole which penetrated the joint and with the drain along the concrete-rock interface.

Hole 18+03 also had a connection to an area outside of the dam. During drilling and water testing of this hole, workmen on the earth embankment foundation preparation crew noted water coming from underneath dental concrete near the downstream termination of the construction joint between Monoliths 30 and 31. The location of the outside leaks was noted and monitored during grouting of this hole. The points of leakage quickly became wet when zone 1 was being grouted with the mechanical packer. Although we started with a 5:1 mix, no cement was noted at the leak locations. Only (apparently) clear water came through. 'Ane hole probably penetrated a thin vertical joint or seam in the foundation which continued into the earth embankment foundation. The seam was apparently wide enough to allow water to pass, but tended to filter out cement particles from the grout. Two quaternary holes drilled 2.5 feet to either side of hole 18+03 took only enough grout to fill the hole volume.

Two mistakes made during grouting should be mentioned. A hole drilled near the Monolith 7/8 construction joint penetrated the edge of the copper water stop in that joint. As soon as the core (containing a semicircle of copper sheeting wrapped around a section of anchor rod) was pulled, the hole

was terminated. After we determined what the drill bit had cut, we grouted the hole with a thick, sanded grout mix and drilled the hole 1 foot farther away from the construction joint.

Hole 15+71 ruptured the 14° pipe which cased the inverted plum bob hole. This hole was planned to go 1 foot to the South Carolina side of the IPB pipe. However, when it was drilled, the bit intersected the pipe tangentially, cutting an elliptical piece of metal from the circumference of the pipe. This was not discovered until grouting of the hole began. As soon as the mistake was discovered, grouting was stopped and the pipe was washed out. The hole was repaired by using a large packer to seal off the IPB pipe and filling the grout hole with a thick, sanded grout mix. Hole 15+71 was redrilled approximately one foot away from this filled, partially penetrating grout hole.

The maximum distance over which hole to hole communication was noted was approximately 100 feet, when grout pumped into hole 0+81 exited hole 1+80 and several intermediate holes during third zone grouting. Communication over 60 feet or more was rare and over 20 feet or more uncommon. Communication between grout holes and other structures include the following (where a structure is called simply a "drain", it is an undrilled drain nipple): hole 0+10 and outside the dam (previously described), hole 0+14 and a porous area in the gutter and gallery floor, 0+66 and the gallery floor, 1+00, a drain, and the M2/3 construction joint, 2+18 and the M4/5 construction joint, 3+39, a drain, and an angled pipe coming up into the gutter on the stairway from a constructed drain between Blocks 7 and 8 (this drain was grouted up), 3+52, the roof of the stairwell, and the upstream face drain closest to the hole, 5+11 and an uplift cell in Block 10, 6+01 with the two closest drains, 8+30 and a drain, 8+40 and a drain, 8+50 and a drain, 9+45 and a drain, 9+65 and a drain, 9+69 and a drain, 15+61 and a drain, 15+71 and an IPB hole (previously described), 16+32 and a drain, 16+80 and an uplift cell in Block 29, 16+93, a constructed drain, and an uplift cell in Monolith 29 (previously described), and 18+03 and outside the dam (previously described).

Table 10 summarizes actual and planned quantities used during concrete dam grouting. It is obvious from this table that less drilling was used than planned, less grout was used than expected, a lower grout placement/drilling ratio than expected was achieved, and the average grout mix was placed at less than a 3:1 w/c ratio. In spite of these things, the integrity of the grout curtain seems to very good. Reasons for this evaluation are the following: exploratory drill holes along the curtain showed no ungrouted open joints, uplift arrays in Monoliths 10 and 23 showed lower pressure downstream behind the grout and drainage curtains during times that water behind the dam was appreciably higher than the Clarks Hill Lake level, split spaced holes usually took little or no grout, and an investigation to determine the reason for the anomalous leakage described above seemed to indicate that the curtain was doing its job. Possible reasons for the grout underrun are: (1) the majority of the weathered, open jointed rock was removed, particularly in the abutments, (2) rock in the spillway and intake sections was not deeply weathered and the top 5-10 feet of rock was removed (5-7 feet below design grade) due to existing (probable) relief joints, and (3) the metadacite exhibits fewer joints than the higher grade gneisses and schists of the Clarks Hill, Hartwell, and West Point Dams

TABLE 10

RICHARD B. RUSSELL

CONCRETE DAM GROUTING

AS OF 1/1/82

(a) ESTIMATED QUANTITIES

GROUT HOLE DRILLING - 46,500 LF
PLACING GROUT - 12,000 CF

GROUT RATIO - 12,000/46,500 = 0.258 CF/LF

(b) ACTUAL QUANTITIES

GROUT HOLE DRILLING - 38,316.5 LF

REDRILL - 8,030 LF

GROUT HOLE DRILLING (ROCK) - 30,286.5 LF

PLACING GROUT - 1269 CF

GROUT RATIO - 1269/30,286.5 = 0.041 CF/LF

DRILLING IS 82% OF BID ESTIMATED QUANTITIES (65% BASED ON ROCK).

GROUTING IS 11% OF BID ESTIMATED QUANTITIES.

(c) PLACEMENT

HOLE SIZE - 1 1/2 INCH OR 0.125 FEET IN DIAMETER

VOLUME OF HOLE - $T \times (0.125/2)^2 \times 30,286.5 = 372.5 \text{ CF}$

AVERAGE MIX USED - 3.5:1 (WATER/CEMENT RATIO)

1269 CF SOLIDS (5076 CF MIX) PLACED IN 372.5 CF OF HOLE OR 372.5 CF SOLIDS WERE USED TO FILL THE HOLE AND 896.5 CF SOLIDS INJECTED INTO THE ROCK.

29.3% OF THE GROUT WAS USED TO FILL THE HOLES. 70.7% OF THE GROUT WAS INJECTED INTO THE ROCK.

upon which the estimate was based. If the spillway foundation had been excavated to original design grade, the open relief joints would have taken a considerable amount of grout and the probable communication and high takes would have required more split spaced holes, increasing the amount of drilling also. Similar considerations apply to the South Carolina abutment. The specifications state that "between Station 9+95.5 and Station 18+57.5, 2-1/2-foot spacing is anticipated within the first zone". The actual condition of the foundation rock was fresh, hard, and unweathered with most joints and seams tight. These tight fractures in the rock would take thin grout mixes, probably filtering out cement particles near the curtain as observed in the outside communication with hole 18+03. If this conjecture is true, the grout remaining in openings near the curtain holes should be more substantial than the 3.5:1 average mix would indicate. In any case, the final verdict on the grout curtain will not be in until filling of the reservoir and time reveals the true results. Additional analysis of curtain grouting is given in Table 9, pages 48 and 49.

10.3-CD <u>Drainage Provisions</u>: Contract Specifications provided for 2-7/8 inch diameter (minimum) holes to be drilled at a 15 degree angle from vertical in a direction downstream and perpendicular to the dam axis. These holes were collared in the downstream corner of the working gallery in small tributary gutters running perpendicular to and ending in the upstream gutters. They were drilled with rotary equipment and diamond bits of NX size to a (vertical) depth of 75 feet on 10-foot centers. Water from these holes runs through the tributary gutters to the upstream gutters and down to sumps in Monoliths 9 and 19 where it is pumped into Clarks Hill Lake.

10.4-CD <u>Dental Concrete and "Broom" Grouting</u>: No dental concrete was used on the Russell dam foundation. The foundation surface over most of the area was irregular and had considerable relief, but the low areas were usually large enough to allow for a "plug" pour prior to mass placement. Where there were narrow depressions in the foundation, concrete inspectors would call for a bucket of concrete with smaller (usually 1-1/2-inch) aggregate and more commentitious material. This bucket would be placed in the depression after application of the basal grout mix.

The basal grout mix used for "broom" grouting on the Russell foundation was composed of 2187 lbs. sand, 629 lbs. cement, 164 lbs. fly ash, 470 lbs. water and 5-1/2 oz. air entraining admixture (per cubic yd.). It was swung into the placement area by gantry cranes (usually) 1-2 yd. at a time. This mix was placed on the wet foundation rock just ahead of concrete placement and "broomed" into any existing cracks and crevices.

10.5-CD Other Treatment and Problems: Problems with flaws in the foundation rock, water, etc., have been dealt with in this report under sections covering the part of the work they affected. The only treatment which has not been covered is the adjustment made due to the abrupt irregularities in foundation topography. Throughout the job, where a stepup face existed, the face was removed if the angle of the face was such as to prevent good vibration of the mass concrete. Also any sharp corners were removed to avoid focal points for stress cracks in the concrete. An additional adjustment made during construction (after recommendation by the District Structural Section) was to use 3-inch, 2000 lb. mass concrete on the foundation and to limit lift height to 3 feet, rather than the usual

7.5, where foundations were very irregular. This was done on the South Carolina abutment monoliths and some in the spillway and intake areas. Prior to this change, a 6-inch, 2000 lb. mass concrete had been used.

10.0 FOUNDATION TREATMENT - EMBANGMENTS

10.1-EE Foundation Grouting: A grout curtain was installed beneath the earth embankments to prevent or slow down seepage; to increase the length of the flow lines through the foundation; and to attempt to fill all open or partially open joints and other openings within the foundation rock. Plates 84 and 85 show the grouting layout in plan view, while Plates 86 thru 92 are profiles of the grout curtain for selected areas. Refer to Table 11 for a summary of earth embankment grouting. In addition to the curtain grouting described in this section, Section 7.1.4-EE describes the consolidation grouting program which was utilized in the diversion channel to consolidate the fault zone encountered there.

Initial grout plans called for a single-line grout curtain parallel to and 10 feet upstream of the dam centerline with grout holes angled downstation at approximately 20 degrees from the vertical. Except between Station 14+20 and Station 18+20, it was anticipated that the grout curtain would extend to a depth of 90 feet (Zone 3) for the primary (set 1 on 20foot centers) holes; 50 feet (Zone 2) for the secondary (set 2, split-spaced on primaries) holes; and 20 feet (Zone 1) for the tertiary (set 3, splitspaced on secondaries) holes. Between Station 14+20 and Station 18+20 (the diversion channel crossing) a fourth zone was added for a total depth of 125 feet for the primary (20-foot centers) holes. Zone 3 (90 feet) holes were on 10-foot spacing and Zones 1 and 2 were on 5-foot spacing. Additionally, the 20 degrees west inclination of the grout holes was waived in lieu of a transition fan to a 20 degrees easterly pattern. This change was necessary for advancing the grout holes into the opposing slope on the Georgia east side of the diversion channel. As the curtain progressed from west to east, the orientation reverted to 20 degrees west once grouting from a relatively "flat" surface resumed (Station 16+20 approximately), see Plate 89.

10.1.1-EE Multiple Grout Lines: Multiple grout lines were used where grout takes for the "A" line so indicated. The second grout line (the "C" line) was placed on the dam centerline and the "B" line was located halfway between the "A" and "C" lines (reference Plates 84 and 85). Holes of adjacent lines were staggered relative to hole stationing on nearby lines. In the event more than three lines were required (as shown on Plate 84, the additional lines were located upstream of the "A" line. In no instances were grout lines permitted downstream of the dam centerline except for the consolidation grouting of the diversion channel fault zone (see Section 7.1.4-EE of this report and Plate 108 for details on consolidation grouting).

10.1.2-EE Grouting Procedures: For a given section of grout line, all primary holes were drilled, pressure washed, tested to Zone 1 depths, and grouted. If grout takes indicated the need for split-spacing, secondary holes were drilled and grouted as for the primary holes. Any additional split-spaced holes were then added and the process repeated until Zone 1 was accepted as being completely grouted. After Zone 1 was accepted, the process was repeated for Zones 2, 3, and 4 in order (i.e., all primary holes

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TABLE 11

RICHARD B. RUSSELL

EARTH EMBANKMENT GROUTING

AS OF 4/1/82

10,584.2/53,066.4 = 0.199 CF/Li

(a) ESTIMATED QUANTITIES

GROUT HOLE DRILLING - 40,000 LF
PLACING GROUT - 5,000 CF

GROUT RATIO - 5,000/40,000 = 0.125 CF/LF

(b) ACTUAL QUANTITIES

GROUT RATIO

GROUT HOLE DRILLING - 81,059.9 LF

REDRILL - 27,993.5 LF

GROUT HOLE DRILLING (ROCK) - 53,066.4 LF

PLACING GROUT - 10,584.2 CF

DRILLING IS 202.6% OF BID ESTIMATED QUANTITIES (133% BASED ON ROCK).

GROUTING IS 211.6% OF BID ESTIMATED QUANTITIES.

(c) PLACEMENT

HOLE SIZE - 2 1/2 INCH OR 0.2083 FEET IN DIAMETER

VOLUME OF HOLE - TT X $(0.2083/2)^2$ X 53,066.4 = 1808.4 CF

AVERAGE MIX USED - 2.5:1 (NATER/CEMENT RATIO)

10,584.2 CF SOLIDS (31,752.5 CF MIX) PLACED IN 1808.4 CF OF HOLE OR 1808.4 CF OF SOLIDS WERE USED TO FILL THE HOLE AND 8775.8 CF OF SOLIDS INJECTED INTO THE ROCK.

17.1% OF THE GROUT WAS USED TO FILL THE HOLES. 82.9% OF THE GROUT WAS INJECTED INTO THE ROCK. within a section of the grout curtain were grouted followed by secondaries, tertiaries, and quaternaries; likewise, Zone 1 was completely grouted before Zone 2, and Zone 2 was grouted before Zone 3.

Grouting always proceeded from the topographic low point to the high point within a section. No drilling nor washing was permitted within 200 feet of a previously grouted hole until after a waiting period of at least 24 hours had elapsed since injection of grout. This requirement was later released to 100 feet without ill effect.

Grout hole drilling was performed with air-track drills using air and water as the drilling fluid. The hole was 2 inches in diameter and was drilled through a 3-inch diameter steel pipe which was seated and grouted into the rock. Four-inch casings were used where grouting through overburden was required. These casings were installed through oversized holes augared to refusal. This proved unsatisfactory as communication was noted when the first zone was drilled for grouting. Later holes in these areas were drilled for a short distance into rock, then stage grouted with a 1:1 mix. This usually prevented communication through the overburden, but it is likely that there was some erosion and possibly some hydrofracturing at the overburden/rock interface during drilling of this first stage. A procedure where casings are drilled through overburden under low pressure and seated in the rock in one step should be used in future contracts. The casings were cut off at the ground surface prior to fill placement.

After drilling and immediately prior to grouting, all grout holes were washed with air and water injected through a pipe which was lowered to the bottom of the hole until the return water was clear and contained no cuttings or joint filling material. After washing, the holes were pressuretested with water at pressures not to exceed 1 psi per foot of hole depth until all return water was clear. For those holes in which the maximum allowable pressure could not be obtained, the holes were washed for 5 minutes with the pumps operating at maximum capacity.

Once all the holes had been washed and pressure-tested within the section to be grouted, grouting operations began. Grout was injected through a packer which was set in . . . inch diameter steel pipe nipple. Pressures going into the grout hole were controlled at the grout header (at the packer, see Plate 112 for grout header arrangement) with grout being circulated continuously. This arrangement prevented the solids from at thing out of suspension and ensured a homogenous grout mix. Likewise, how accurate control of the down-hole pressures was obtained.

Grout mixes ranged from 0.6:1 water to cement ratio (by volume) to 5.0:1 water to cement ratio with 2% bentonite (.02 x cement wt.). Normally, grouting began with the thinnest mix (5:1) and was thickened only if grout takes indicated the need for such thicker mixes, so as to obtain refusal. On some grout holes with abnormally large water takes during washing and testing, a 3:1 grout mix was called for initially rather than the 5:1 mix which was the "normal" initial mix.

Refusal criteria for a given hole was considered met if less than one cubic foot of grout at the maximum allowable header pressure could be pumped into the hole in a ten minute time period. Upon obtaining refusal for a

zone, the grout was allowed to achieve its initial set and the hole was then washed out (except when takes were higher than normal) so as to preclude the requirement for redrilling that particular zone. If the hole was complete after reaching refusal, washing of the hole was not performed, but rather the hole was backfilled completely with grout and allowed to set. Upon completion of a hole, the steel pipe nipple was either removed or cut off even with the ground surface. For a graphic description of general grouting procedures refer to Plates 110 and 111.

10.1.3-EE Effectiveness of Curtain Grouting: A visual inspection of the downstream embankment toe areas could be indicative of the success of the grout curtain. The absence of unusual seepage on the downstream sides of the embankments can be attributed at least in part to a comprehensive grouting program. More information regarding the grout curtain's effectiveness will be made available when the Embankment Criteria and Performance Report is completed.

10.2-EE Foundation Compaction and Consolidation: Consolidation of the hanging wall of the diversion channel fault was achieved by grouting. This grouting operation is described in Section 7.1.4-EE of this report.

10.3-EZ Dental Concrete: Dental concrete was used to fill narrow depressions formed by blasting, ripping, or dental excavation, and against vertical or overhanging rock ledges. Dental excavation was performed on all deeply weathered or extremely shattered rock (primarily vertical mafic dikes) using hand tools and/or loader-backhoes. A Cat 245 hydraulic excavator was used in the diversion channel for dental excavation of the fault. The depth of excavation criteria used was generally to excavate to a depth equal to two times the width of the excavation. The concrete used was a 2500 psi ready-mix which was vibrated and wood trowel finished. A lowslump mix was used for the fillets and a slightly higher slump was specified for dike excavations. Forms were not normally required, but a toe board was used for fillets over about three feet in height. Section 7.5-EE of this report describes a large (244 cubic yards of concrete) fillet which was required in the South Carolina embankment cutoff trench. Contract quantities for both dental excavation and dental concrete were entirely too low and resulted in large overruns (see Appendix G, Variations In Estimated Quantities). For further information about locations that required dental concrete placement refer to the Embankment Criteria and Performance Report.

10.4-EE Fibrous Shotcrete Treatment: Upon completion of all other forms of foundation treatment, a fibrous reinforced shotcrete was applied to the surface of all firm and fractured sound rock. The intent was to minimize the possibility of the highly erodible impervious core material from being transported into openings in the foundation rock. The material was a dry-mix of sand, cement, and steel fibers applied pneumatically with water added at the nozzle. Specifications called for a minimum of 3 inches of shotcrete; any test revealing less would require an additional 3 inches at the contractor's expense. There was no specified restriction on maximum thickness. In generally flat areas, the measured thickness was normally 3-1/4 to 3-1/2 inches (contractor used 3-inch long nails as thickness guides). However, in irregular surface areas with abrupt changes, thicknesses on the order of 6 to 8 inches were not uncommon in one spot in order to obtain 3 inches on an adjacent higher spot. Consequently, many abrupt details were eliminated through the use of shotcrete. Many of these areas would have

needed slush-grouting or mortar treatment if shotcrete had not been applied. Test beams were made and cured in the field. Further information about areas that were shotcreted will be included in the Embankment Criteria and Performance Report.

Testing of the shotcrete product was specified to be through the use of panals, beams, and cores. Only the cores sampled the in-place shotcrete. It was dytermined after a few days that the panels were of very little benefit on a daily basis once a nozzle operator had been approved. It was more desirable to obtain core samples to monitor thickness and consistency. Production rates were initially only 1 to 2 cores per week. Therefore, daily panels were discontinued and daily cores substituted. Of the 116 cores drilled, only 1 revealed less than the required 3" (2" core recovered, this area re-shotcreted).

Laboratory testing of the shotcrete was performed at the South Atlantic Division Laboratory. Shotcrete beams and cores were selected for testing after field curing. Normally, ages of 7, 14, and 28 days were selected, although other ages were occasionally tested. Beams were tested in flexure and in unconfined compression. Tables showing shotcrete test results are contained in Appendices D thru F.

10.0-PH FOUNDATION TREATMENT - POWERHOUSE

10.1-PH Rowerhouse Foundation Grouting: Unlike other grout programs at the Russell Project, the powerhouse foundation was grouted for the purpose of foundation consolidation rather than for control of subsurface flows. This being the case, grouting was performed in only one stage with all holes being the same depth. The distribution and locations of the grout holes are shown on Plate 125 and Table 12 lists pertinent data.

The physical location of the grouting was downstream of the service bays and between the draft tubes of each unit. After rock excavation for the unwatering trench, these areas formed "noses" or protrusions of rock which were eventually encased by concrete. The rationale behind leaving these protrusions rather than excavating them and backfilling with concrete was based upon an engineering economics analysis which concluded that it would be more cost effective to leave the rock and grout rather than remove the rock and replace it with concrete. The rationale for grouting the rock was based upon the fact that the rock, rather than being a homogeneous mass, consisted of a finite number of discrete joint blocks with varying amounts of point contact, and since these areas would be subject to vibration from the generator units it was deemed advisable to consolidate them so that their behavior would approach that of a homogeneous mass.

To achieve this homogeneity, it was determined that a portland-cement based, non-shrink, non-expansive grout would be satisfactory from the standpoint of both economy and effectiveness. The contract specified that the grout would consist only of water, Type I portland cement, and unpolished aluminum powder and that the proportions would be designated by the Contracting Officer. This wording of the specifications proved to be very fortuitous as it allowed project personnel considerable leeway in adapting the grout mix to the conditions encountered.

Observations during grouting operations together with intimate knowledge of the geologic characteristics of the project site lends to the conclusion

TABLE 12

TABLE OF GROUT DATA FOR POWERHOUSE

No. of Holes (each 30.0 feet deep)

Linear Footage 5310 feet

Total Grout Take 374.72 ft.3

Lowest Take/Foot 0.0 ft.3

Highest Take/Foot 1.23 ft.3

Mean Take/Foot 0.0705 ft.3

Standard Deviation 0.1492

Total Grout Take as Percentage of Jovernment Estimate - 11.5

that the grout program objectives have been fully met. The success of the program was largely due to the manner in which the specifications were written (see preceding paragraph for summary) and it is suggested that careful consideration be given to the idea of writing future grout specifications in a similar fashion as opposed to specifying X number of bags of a specified mix before being allowed to switch to a new grout mix. It is felt that design objectives can often be met a lower cost and in less time if the field personnel are provided with flexible and non-confining specifications.

10.2-PH Foundation Drainage: Foundation drains in the powerhouse structure are designed to intercept and divert any subsurface waters whose hydraulic potential is great enough to pose hydrostatic uplift problems. The foundation drainage system is composed of three main types of elements. The first of these is the system of relief wells drilled into rock underneath the service bay and draft tube floor slabs. Plate 126 depicts the location of these drains while Plate 129 depicts typical profiles. These drains are 3-inch diameter holes drilled forty feet into rock and are designed to intercept, at depth, any groundwater which may be influenced by either Clarks Hill or Richard B. Russell impoundments.

The second type of drainage element utilized in the powerhouse is a system of weep holes which extend through the draft tube floor slabs and are connected to collector boxes which are installed at the concrete/rock interface. The collector boxes are half-round corrugated PVC, concave up, which were installed over open joints in the foundation rock. To prevent dislocation during concrete placement, these boxes were pinned to the rock with short lengths of re-bar and sealed against intrusion of concrete by cement mortar. Plates 126 and 129 show the locations of the weep holes and typical sections of the collector box system.

The third drainage element is located in the erection bay substructure wall and the tailrace hang-on wall. Weep holes through the tailrace wall and a lateral drain along the erection bay wall were designed to limit the backfill saturation level to three feet above tailwater for normal conditions.

10.3-PH French Drains: In those areas of the powerhouse foundation where in-flows of water occurred, french drains were installed prior to placement of concrete. The riser pipes were extended as high as necessary to overcome hydrostatic head and when concrete had been placed to this level, the drains were grouted to refusal (when only one riser was utilized) or until grout return through the second riser (in the case of extensive linear in-flow zones) was achieved. The majority of the french drains were installed in the bottom of the unwatering trench where sub-horizontal relief joints day-lighted in the walls of the trench. In those cases where only a vertical large-diameter pipe was sufficient to isolate the in-flow, a sump pump was used to control the in-flowing water until concrete had been placed above the piezometric surface. Once this was accomplished, a small diameter grout pipe was installed in the drain, the pump was removed and the large pipe was backfilled with a clean crushed stone and grout was pumped into the drain until all water had been displaced and grout return at the surface achieved.

- 10.4-PH Foundation Preparation: Pry bars, wedges, jack hammers, and water jets were used in preparing the foundation for concrete. All loose and drummy rock, rock underlain by clay seams, overhanging rock ledges, and excessive staining and mineral deposits were removed under the direction and supervision of the Project Geologists who had final authority regarding the suitability of the foundation. Most areas of the foundation required extensive rock removal due to the intensive jointing, the presence of numerous shear zones, and the effects of blasting during initial excavation. Once the foundation satisfied the Project Geologists, the areas were photographed (see Appendix K) and mapped and the foundation was approved for rebar and concrete. Any excessive delays between foundation approval and concrete placement required a second inspection and approval by the Geologists immediately prior to placement.
- 10.5-PH Foundation Mapping and Approval: The entire powerhouse foundation area was mapped by the Project Geologists. Field mapping was at the scale of 1:60 with the final map (Plates 120-123) at a 1:240 scale. Due to the nature of the construction sequence, the largest area available for mapping at any one time was the service bay areas and so it was considered that the use of a plane table and alidade was unnecessary. Instead, each area was laid out in a 5-foot grid and geologic features were located relative to the intersections of the grid. Among those features plotted on the maps were rock type, contacts, discontinuities, mineralization, and any other features which were deemed to be of importance in the event of any future foundation problems. Before mapping commenced, the foundation was examined by a Geologist and only after the foundation was considered to be ready for concrete was it mapped. Foundation approval and mapping was conducted from December 1981 through August 1983 when the last of the rock was finally covered by concrete. In addition, mapping of the tailrace slope was accomplished in 1984, immediately prior to inundation.

11.0 FOUNDATION INSTRUMENTATION - CONCRETE DAM

11.1 Concrete Dam Structures: The three types of structures in the concrete dam are: non-overflow, power-intake, and a gated spillway section. Bearing pressure on the foundation is less than 15 tons per square foot. Sliding or shearing stress on the structures is less than 100 psi. These loads are well within the capacity of the foundation (refer to Section 3.5 for more details).

Plates 53 thru 56 show the design details and locations of instruments that were planned for the dam. Plate 57 shows final locations of instruments. Foundation instrumentation installed in the concrete dam includes: Uplift cells in Monoliths 7, 10, 23, 26, 27, 28, and 29, side by side plumb bobs and inverted plumb bobs in Monoliths 16 (inverted plumb bob is 100 feet into the foundation) and 26 (inverted plumb bob is 85 feet into foundation), and deflectometers (inclinometers) in Monoliths 27 and 28.

- 11.2-CD <u>Purposes</u>: Instrumentation was implemented on this project in order to maintain data on: (1) hydrostatic uplift on the foundation under several of the concrete dam monoliths, (2) deflection and tilting of individual monoliths, (3) rates of inflow of water into the inspection gallery, and (4) seismic activity.
- 11.3-CD Uplift Cells: The locations of uplift cells in the foundation were based on a random sample of the various type of structures above the foundation and variations in elevations and conditions of the foundation. The blocks and types of structures that were selected are as follows:
 - Block 7 Georgia non-overflow.
 - Block 10 Power intake section.
 - Block 23 Typical spillway section
 - Block 26 End block of spillway
 - Block 27 South Carolina non-overflow.
 - Blocks 28 and 29 Variation in rock condition under South Carolina non-overflow.

Uplift cells were located from the upstream edges of the blocks to the downstream edges of the blocks in order to determine the variations in uplift pressure from upstream to downstream. Uplift cells were also located both upstream and downstream of the foundation drains to determine the effectiveness of the drains. The uplift cells were place on top of rock, 4 feet deep and 10 feet deep in order to monitor the uplift onditions at the contact plane and various levels below the foundation surface.

11.4-CD <u>Plumb Bobs</u>: Plumb bobs were installed in spillway end blocks to measure tilting of the dam under reservoir loads. Inverted plumb bobs were also installed to determine if the pool and load from the concrete dam effected the foundation. The measurement of the plumb bobs are compared

against the alignment surveys to verify the accuracy of the surveys. The data is used to evaluate the safety of the structure.

- 11.5-CD <u>Inclinometers</u>: Inclinometers were installed in Blocks 28 and 29 for the same purposes as the plumb bobs. Inclinometers are considerably less expensive than the plumb bobs, but are at the same time slightly less accurate.
- 11.6-CD <u>Seismic Instrumentation</u>: Earthquake instrumentation consists of two strong-motion accelerographs, one Wilmot Seismoscope, and one peak-recording accelerograph.

The strong motion accelerographs are activated by strong locally occurring seismic activity. The data is recorded on 70 MM film stored in a magazine inside the instrument. The unit is battery operated and its triggering mechanism is electrically activated.

11.0-EE INSTRUMENTATION - EMBANKMENTS

Instrumentation prescribed for the earth embankments included opensystem type piezometers, gas-operated piezometers, settlement plates, inclinometer / settlement gages, surface reference monuments, and benchmarks. Refer to Plates 113 and 114 for design locations of these monitoring devices. Plate 115 contains installation and functional details.

The as-built locations and descriptions of the instruments actually installed in the project (which differed somewhat from designed locations and descriptions) will be assimilated into the Earth Embankment Criteria and Performance Report.

12.0-CD POSSIBLE FUTURE PROBLEMS - CONCRETE DAM

12.1-CD Conditions That Could Produce Problems:

- a. Relief joints in the Georgia abutment These are described in previous sections. They are clean, irregular, and show good rock to rock contact where exposed, so they were not considered to pose any structural danger to the dam. Any voids should have been filled by grout (communication was evident along these joints during grouting operations). However, any geologist examining the Russell Dam foundation should be aware of their presence. If portions of these extensive planes were incompletely grouted, they could provide avenues for seepage.
- b. The Monolith 28/29 step-up face Block 28 was excavated much deeper than originally designed due to weathered and deteriorated rock. A dike swarm/shear zone with evidence of multiple movement episodes cuts the face. A near horizontal seam with clay and mud filling terminated irregularly in Blocks 28 and 29. Triangular remnants of this seam still exist in the upstream and downstream corners of the Block 29 foundation on the west side of the block. The grouting plan here was altered to cut off flow at the face. However, the area should be monitored for excess seepage.
- 12.2-CD Recommended Observations: The areas mentioned above should be monitored for seepage. Uplift cells in Monolith 7 should be helpful for observation of seepage in the Georgia abutment, along with the drains downstream of the grout curtain. The foundations in the Monolith 28/29 area can be monitored by uplift cells in both blocks and in Blocks 26 and 27. Drains downstream in these blocks and horizontal drains in the Monolith 29 face can also be observed.
- 12.3-CD Lessons Learned: Some effort should be made to require the contractor who blasts rock to bear some of the burden of overblasting. In order to do this, the point at which the work is split into different contracts should be changed from that used on this dam. A possible better point might be at 5 feet above design grade. The excavation contract could be stopped at 5 feet above grade and the concrete construction contractor be required to remove the last 5 feet of rock and to prepare the surface and place concrete. This would also have the advantage of allowing for minor changes in design grade between inspection of excavation and bidding on the construction contract. Alternatively, the excavation contractor could be required to do all excavation and foundation preparation, then to place concrete to 5 feet above design grade. Both of these changes would prevent the problem of having a prepared rock surface spall and loosen in the time between the presentation of the two contracts and would minimize claims and other problems which arise when two contractors have to operate in the same area.

Another possible improvement in current Corps practice is the drilling of probe borings as well as core borings. In locations where the top of rock is extremely irregular such as the Russell area, it is difficult to delineate pinnacles of rock and pockets of soil. Until geophysical methods become discriminating enough to do this, it might be economically beneficial to use a self-propelled drill rig such as an air-trac drill to probe to the top of rock. With a small diameter bit, drilling through overburden to the

top of rock only, a high production rate and low unit cost could be maintained. For every foot of NX core hole eliminated, at least 10 feet of air-trac hole could be drilled. This would allow denser coverage, greatly increasing the chances of weathered linear features such as faults being discovered, and preventing material overruns such as that which occurred on Class I excavation on the Richard B. Russell Dam. The usual splitspoon and coring holes could be drilled to determine type, nature, and engineering properties of materials and suitability of a specific site. Then a pattern of probe borings could be made to determine top of rock topography and depth of overburden. Finally, additional core borings could be made to explore low spots and linear features which showed up in the probe boring pattern. Over 400 borings were drilled for Line C along which the Russell Dam was built. This expense could probably have been lowered, or at least more information to guide design could have been gathered for the same price.

12.0-EE POSSIBLE FUTURE PROBLEMS - EMBANKMENTS

- 12.1-EE General: The potential for future foundation problems which might constitute a threat to the integrity of the embankment appears to be very low. The probability of foundation conditions developing which would create maintenance or aesthetic problems is not so remote.
- 12.2-EE Quarry Shear Zone: The only foundation area with a high probability of developing into a problem which will require remedial measures in the future is the Dravo-Groves Quarry shear zone. The Dravo-Groves Quarry was operated by the concrete dam contractor as a source of aggregate and other processed stone for the Richard B. Russell Project and is located within the reservoir area near the South Carolina shore. Plate 3 indicates the location of the quarry relative to other features of the project. Investigation by Government Geologists of a zone of sheared and brecciated rock exposed by quarry operations resulted in the hypothesis that this shear zone extends, in the subsurface, through the South Carolina Saddle Dike foundation. The saddle dike was not founded on rock and no grout curtain was installed under the dike.

Evaluation of the geologic investigation resulted in the decision that no foundation improvement would be performed under the construction contracts. This decision was based upon the supposition that if any remedial work were required that it would be more economical to do so in the future than to immediately engage in a large-scale foundation improvement program when the need for such work was not evident. A system of weirs has been installed in the ravines downstream of the saddle dike and flow rates are being monitored regularly by Operations Division. If the flow rates begin to increase over time, Engineering Division is prepared to initiate an exploration and investigation program and to recommend whatever remedial work (if any) is required.

12.0-PH POSSIBLE FUTURE PROBLEMS - POWERHOUSE

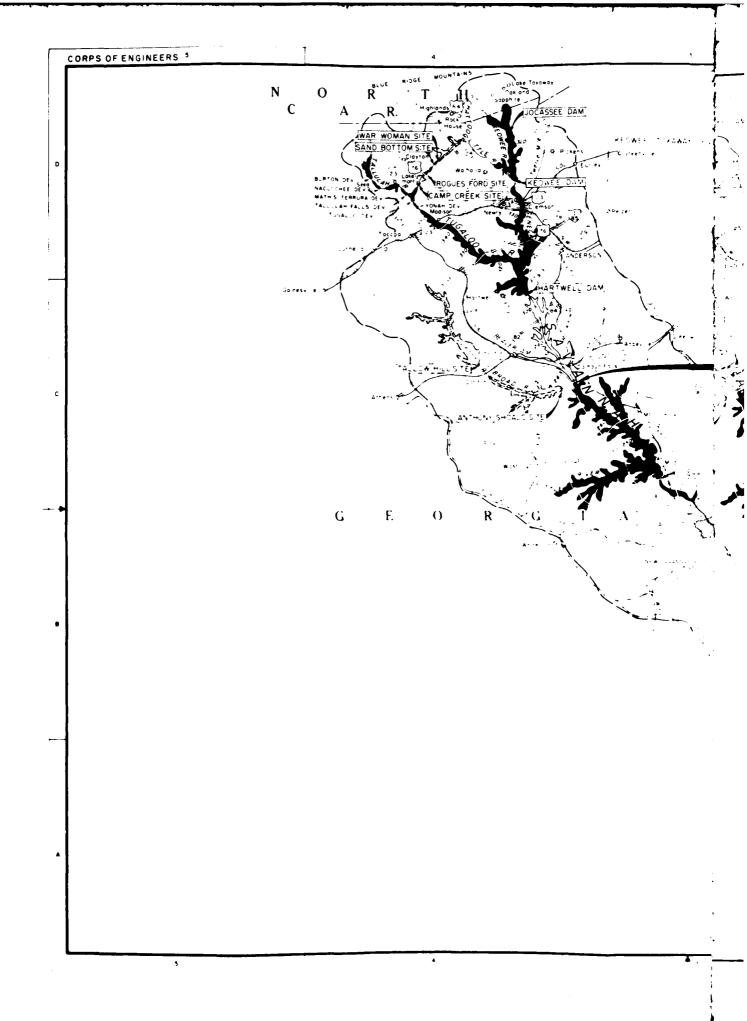
The likelihood of any major problems developing in the powerhouse foundation appears to be very remote. The only foreseeable problems would be that some of the relief wells might become clogged for some reason and it is felt that this would not create a serious problem as the entire structure is well anchored with an extensive network of rock bolts. Also, if the

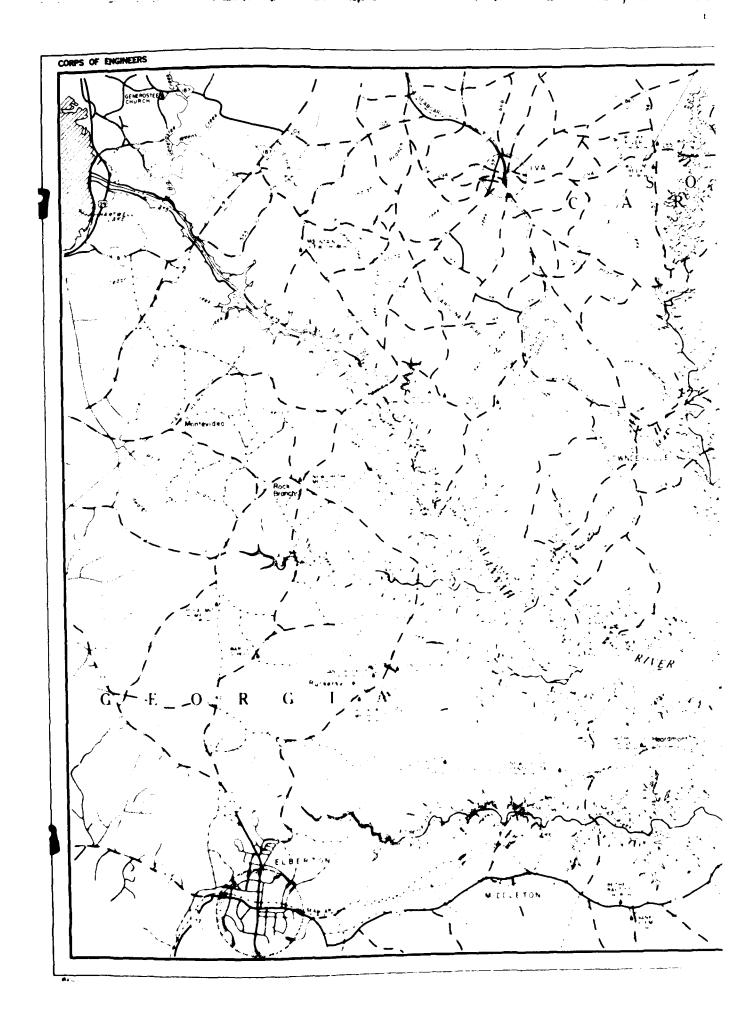
wells do become clogged, it would be a relatively minor operation to either ream the existing wells or to drill new wells.

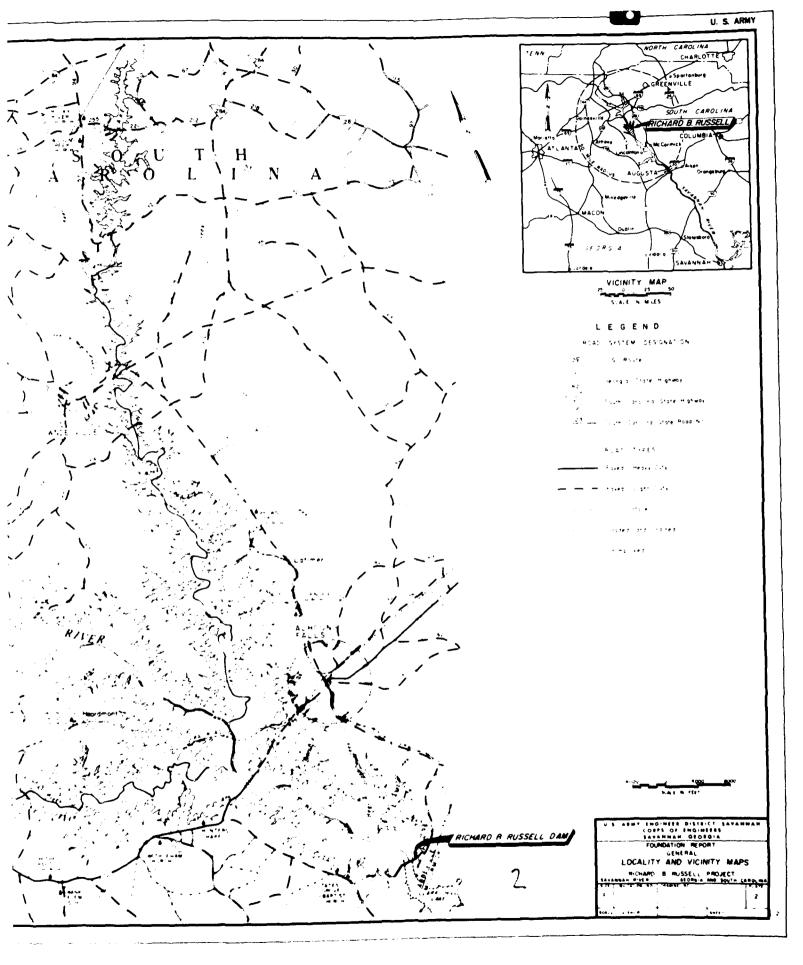
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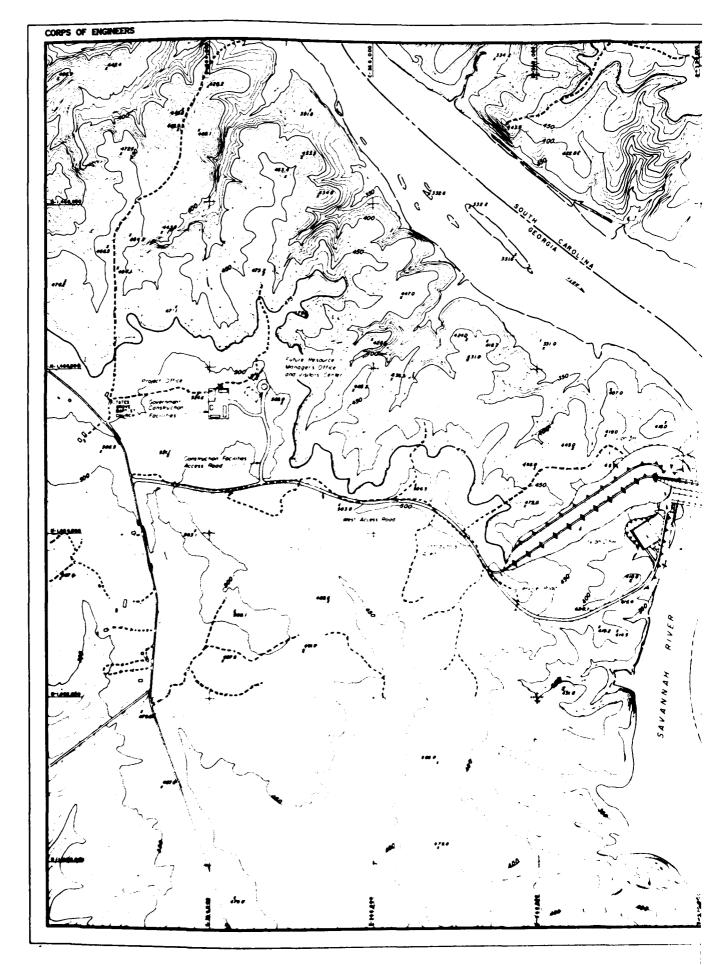
PLATES

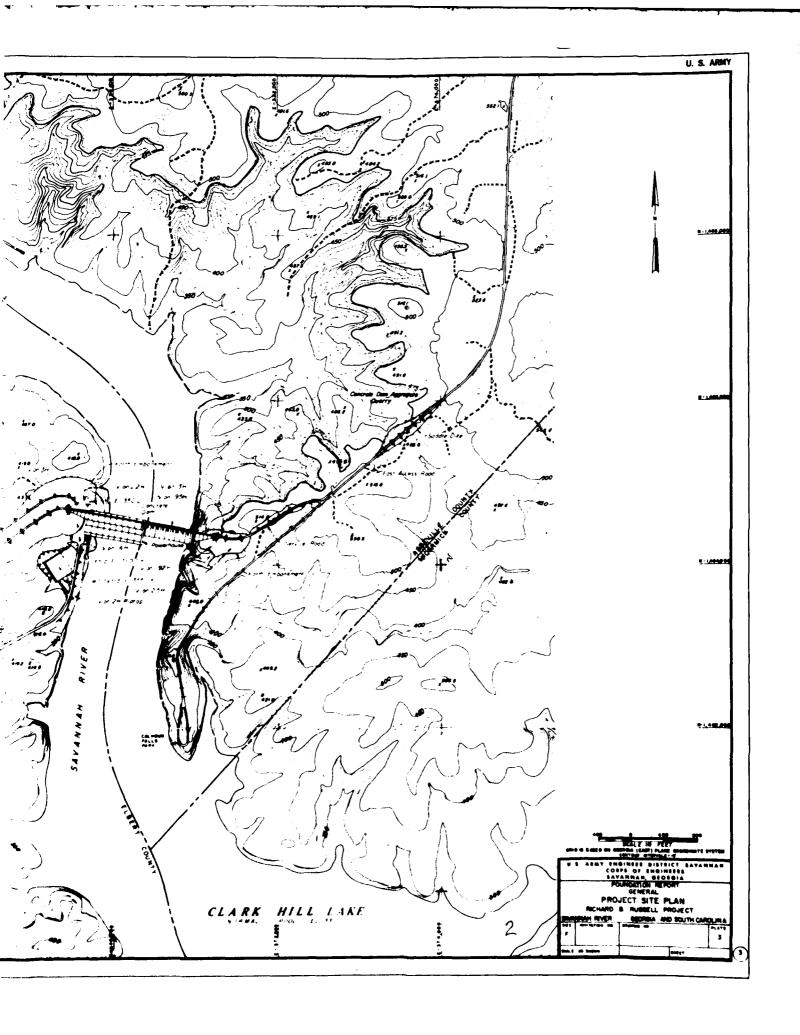


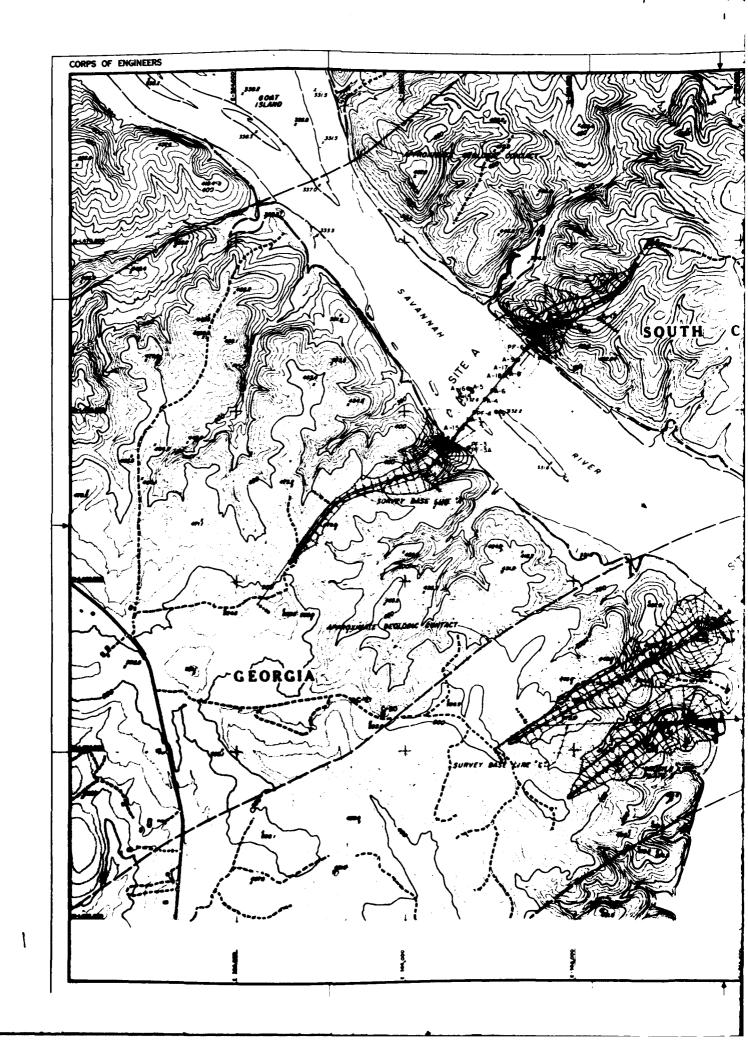


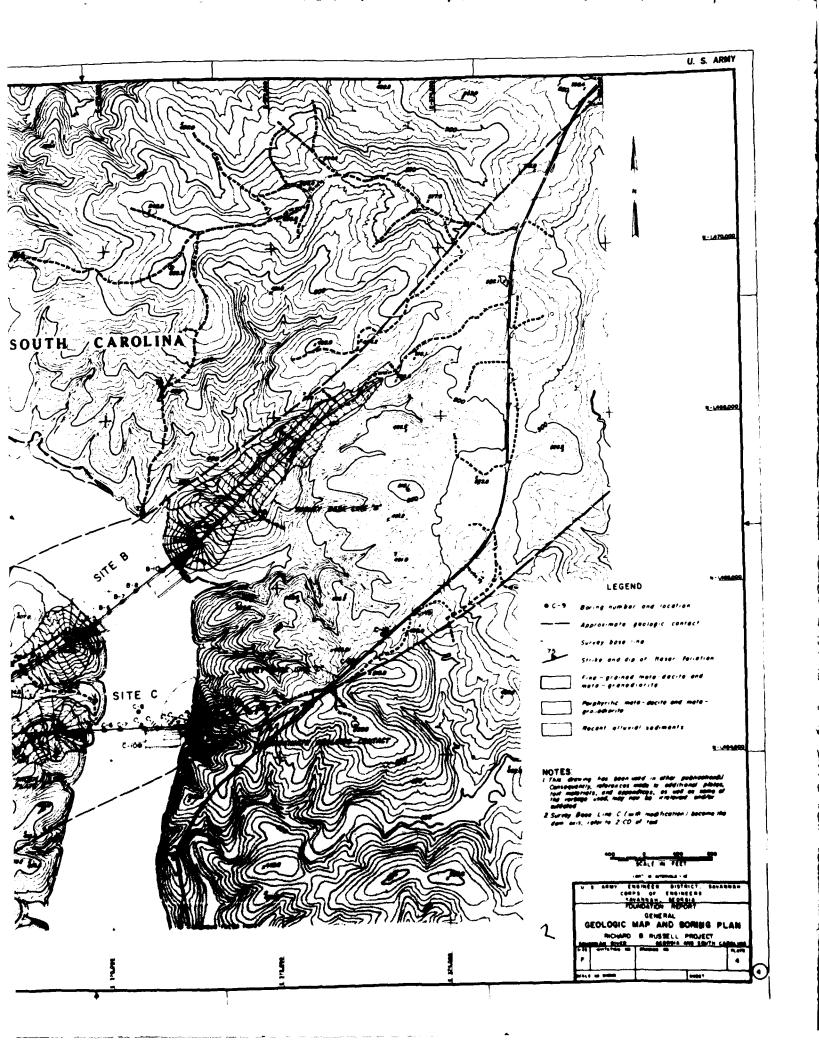


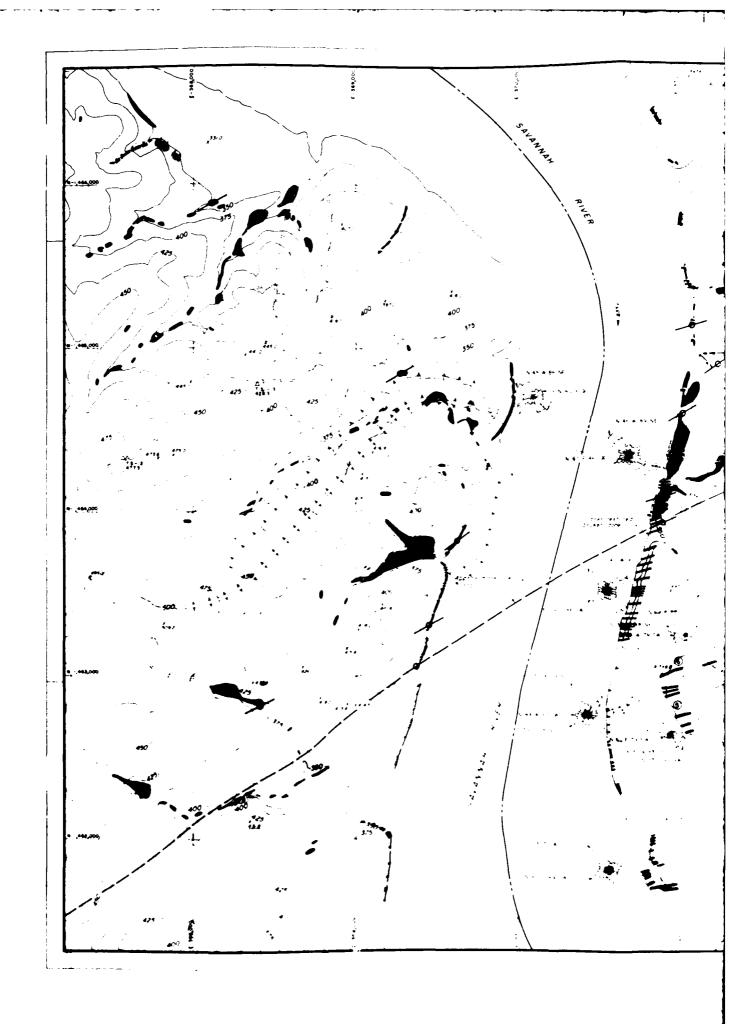
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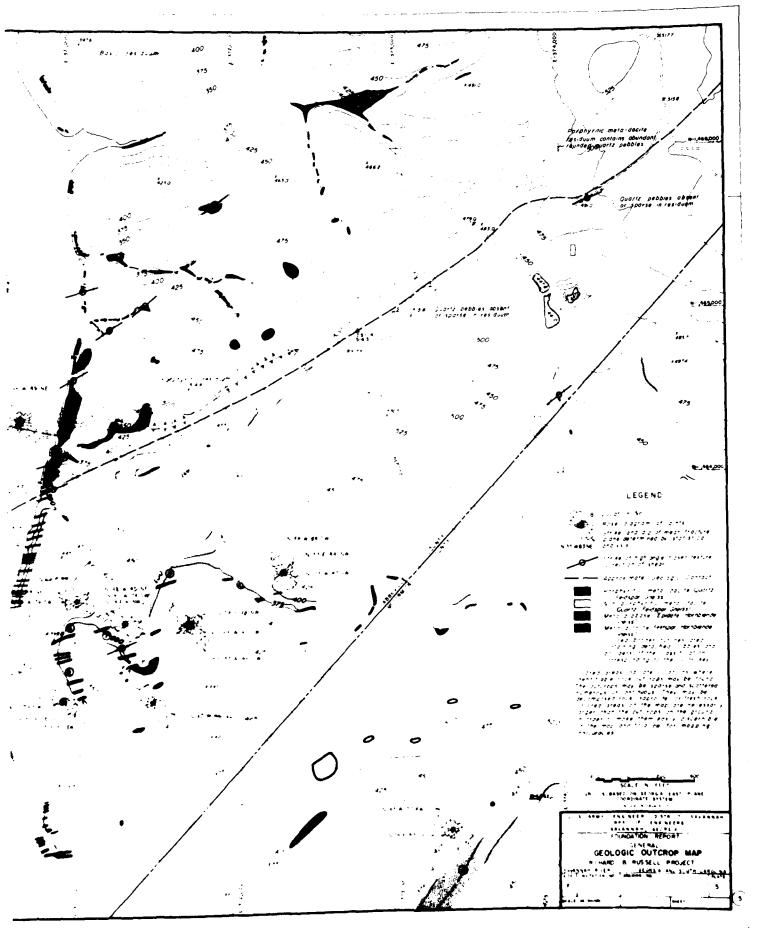


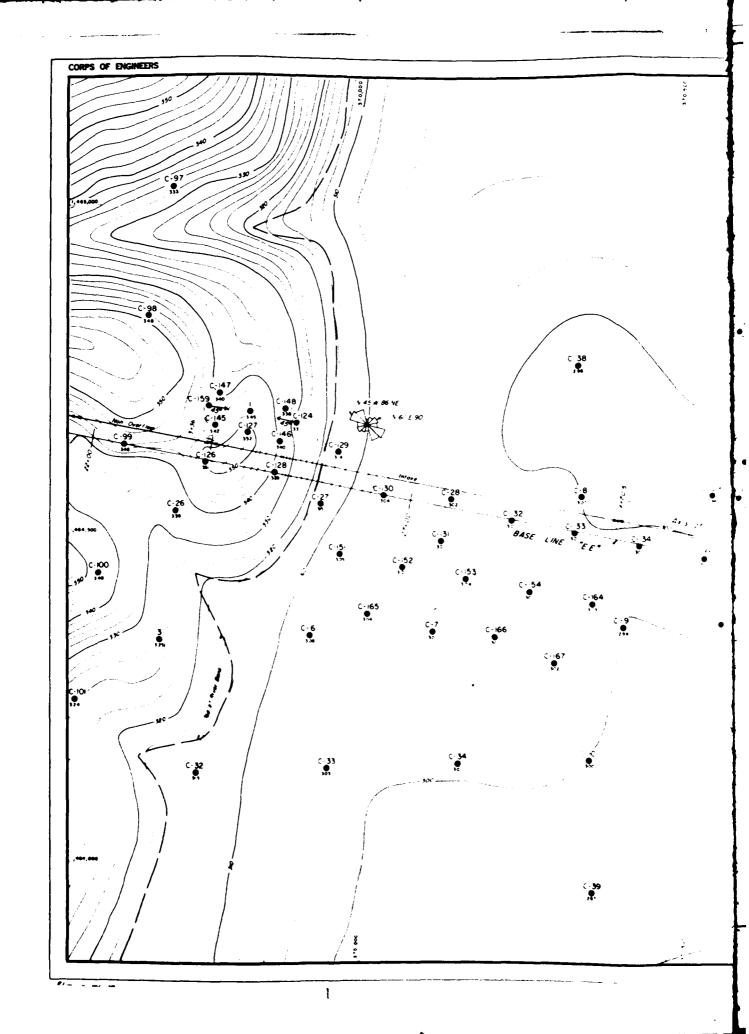


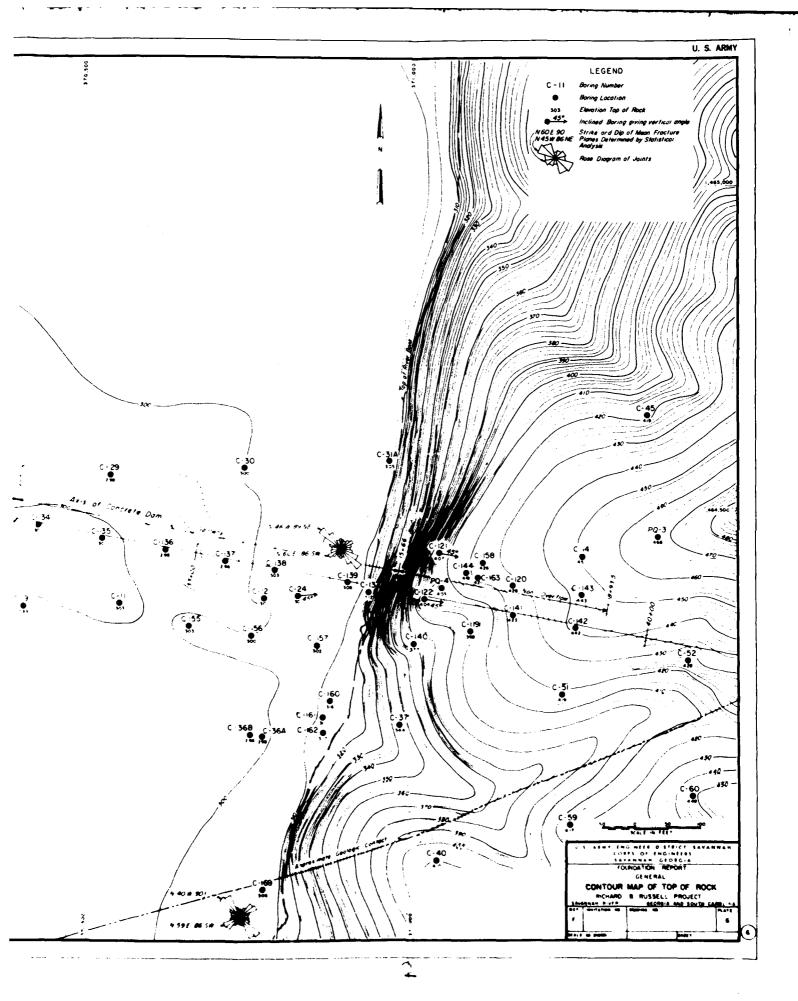


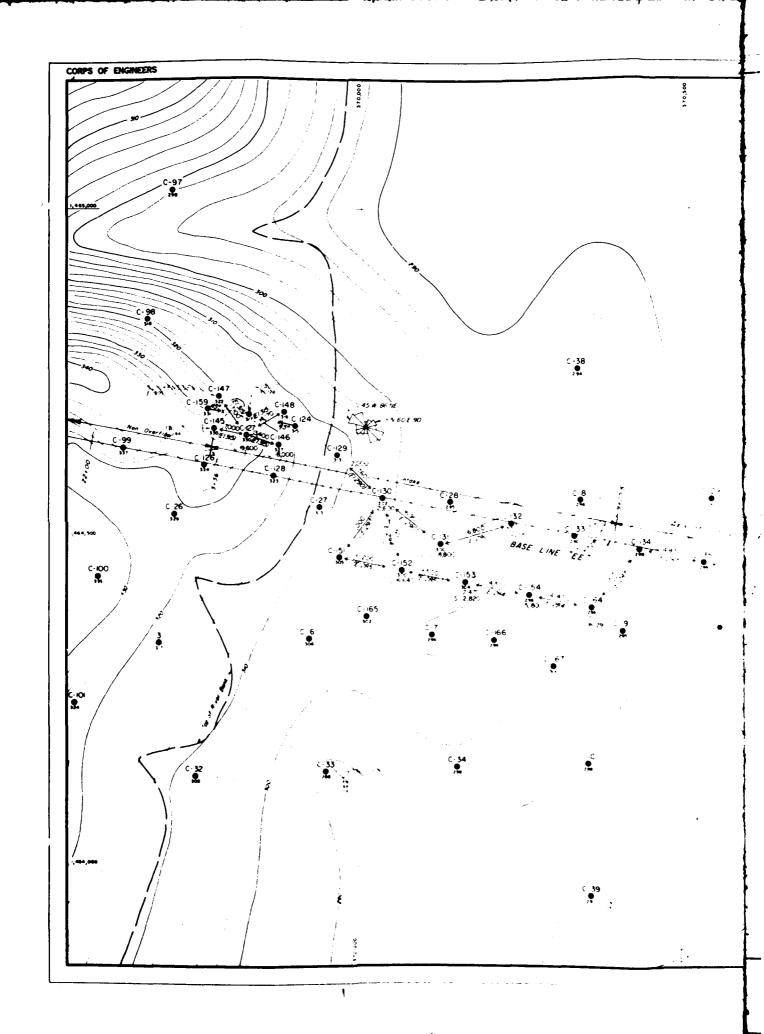


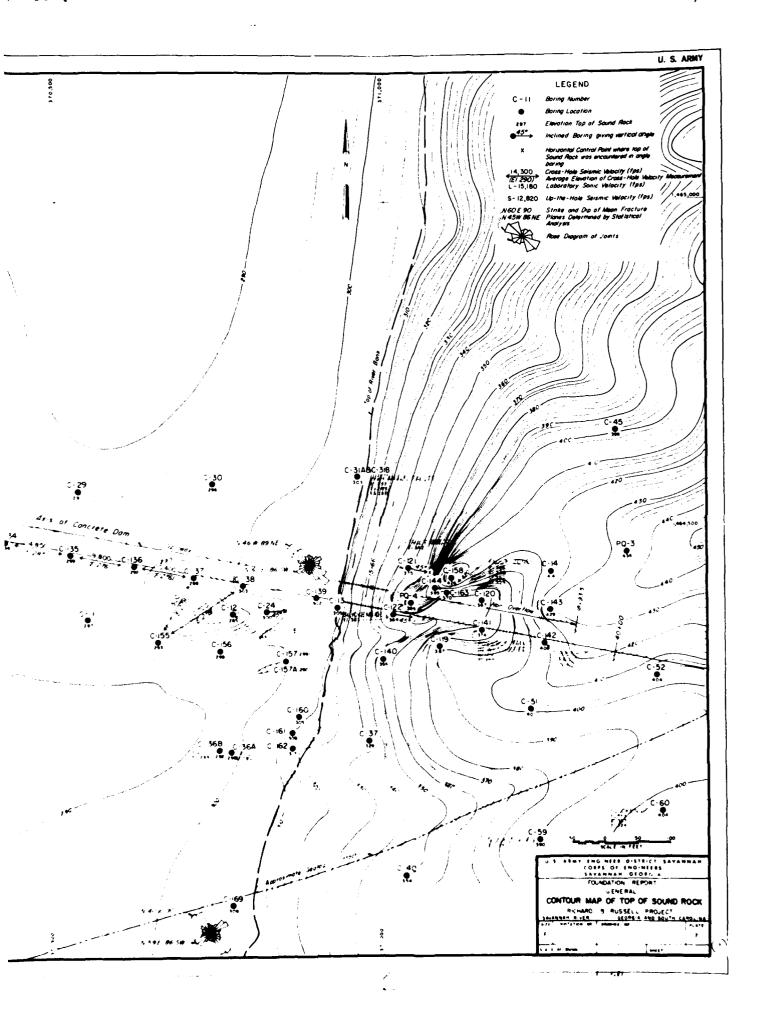


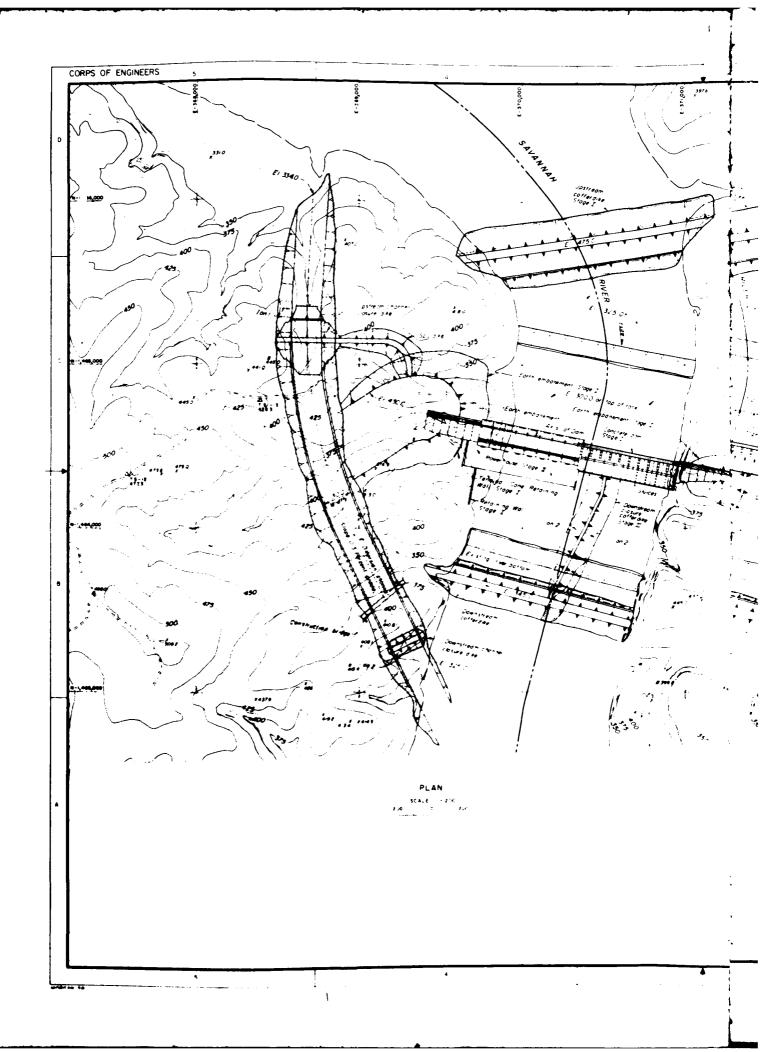


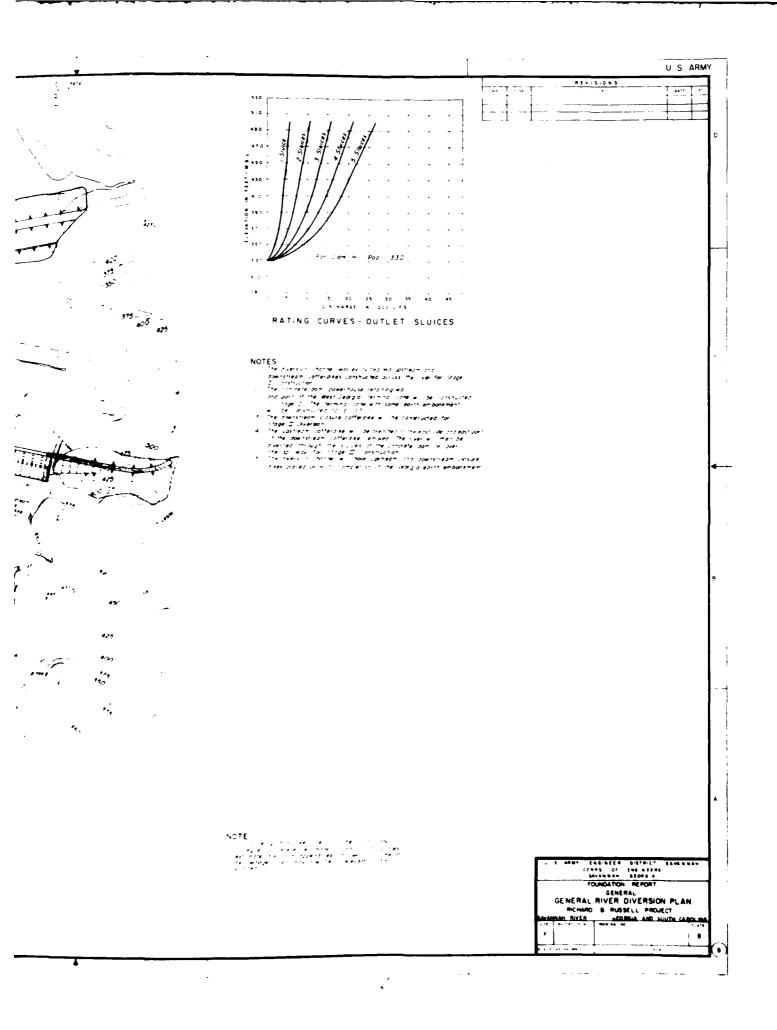


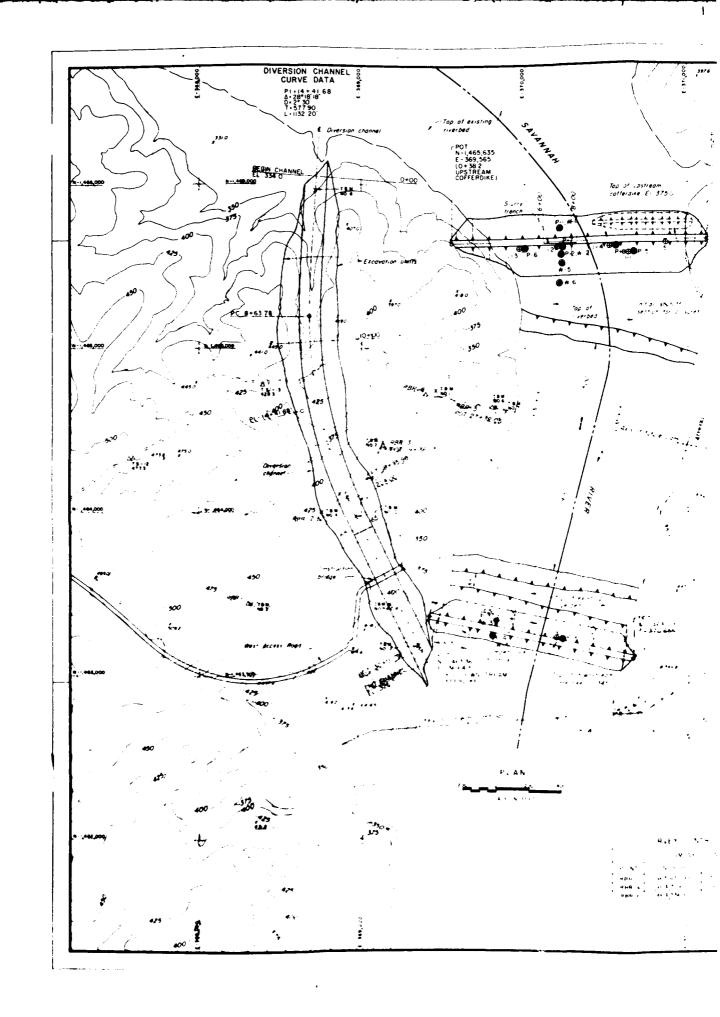




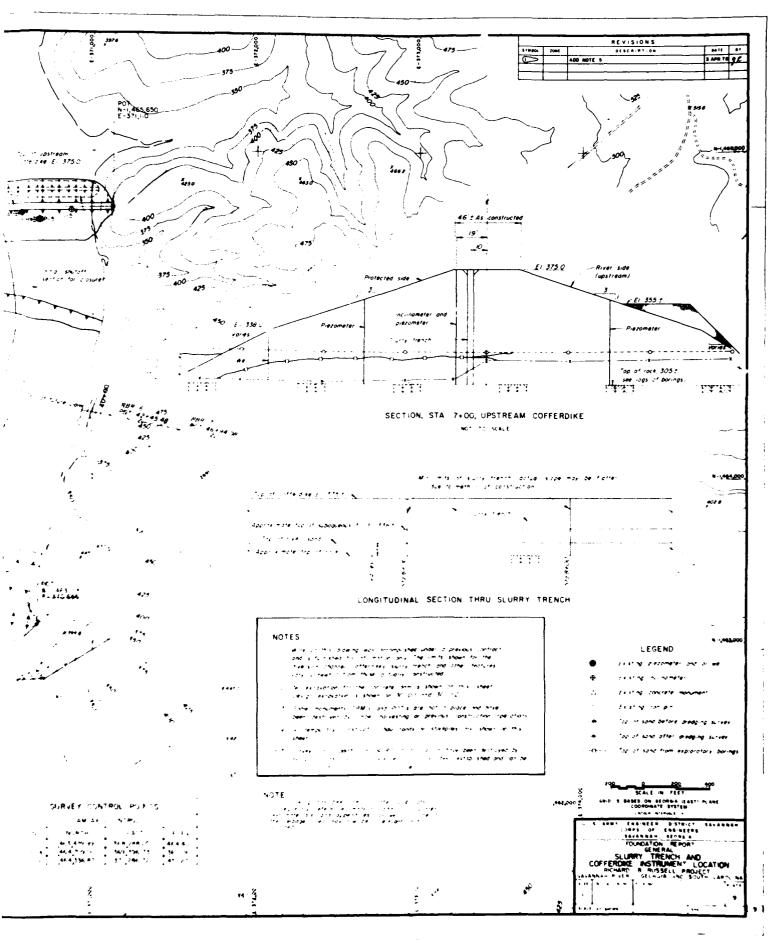


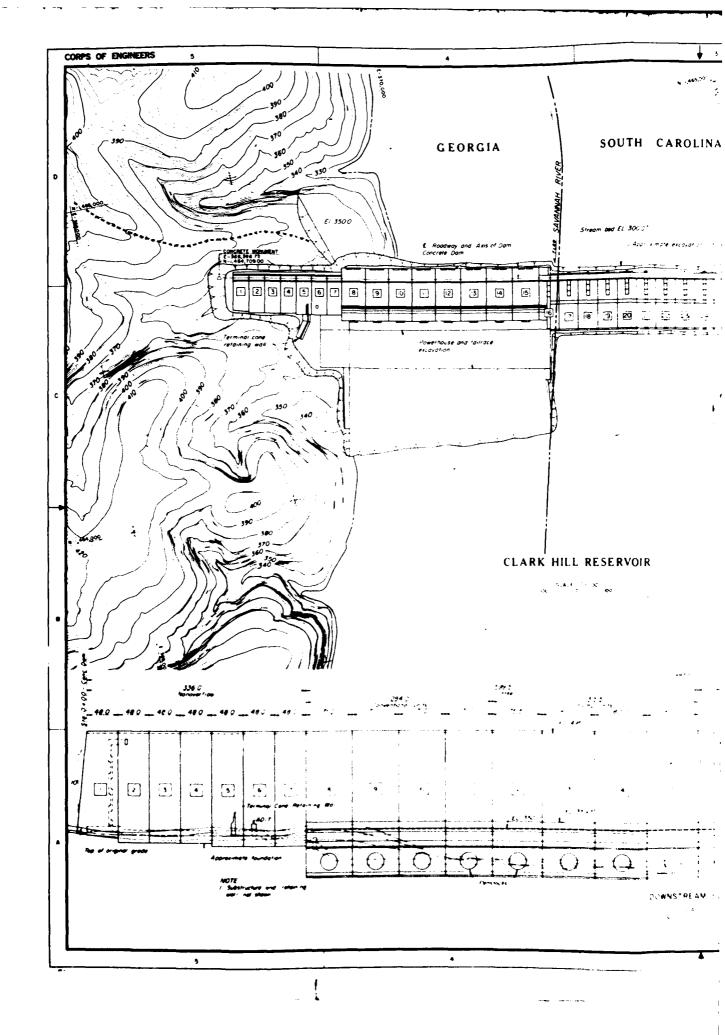


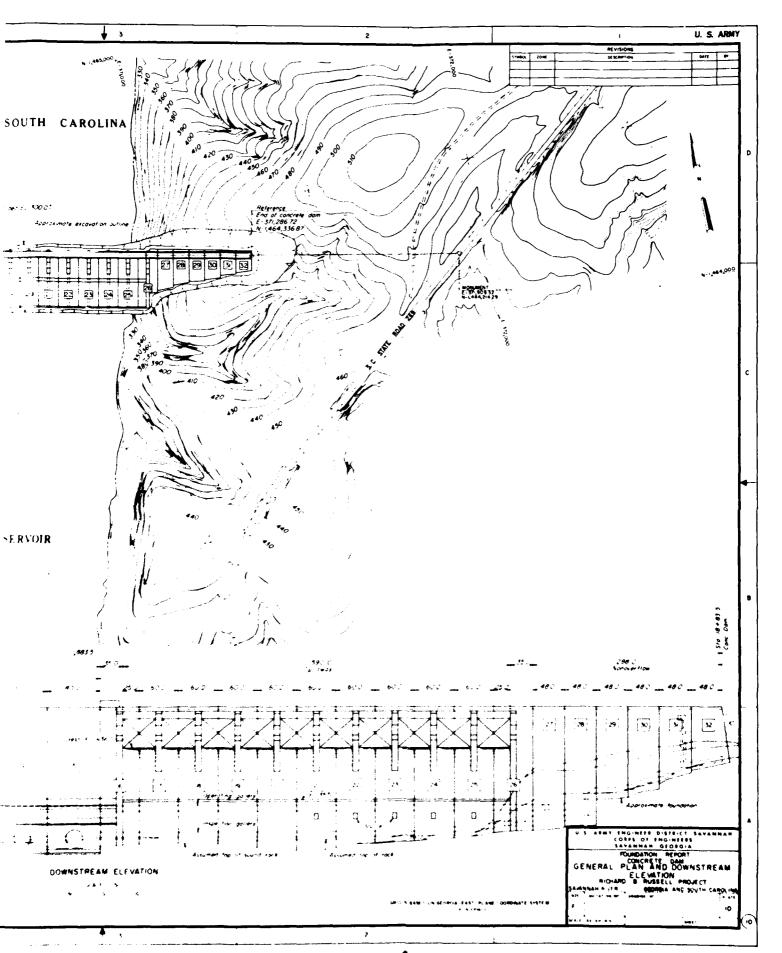




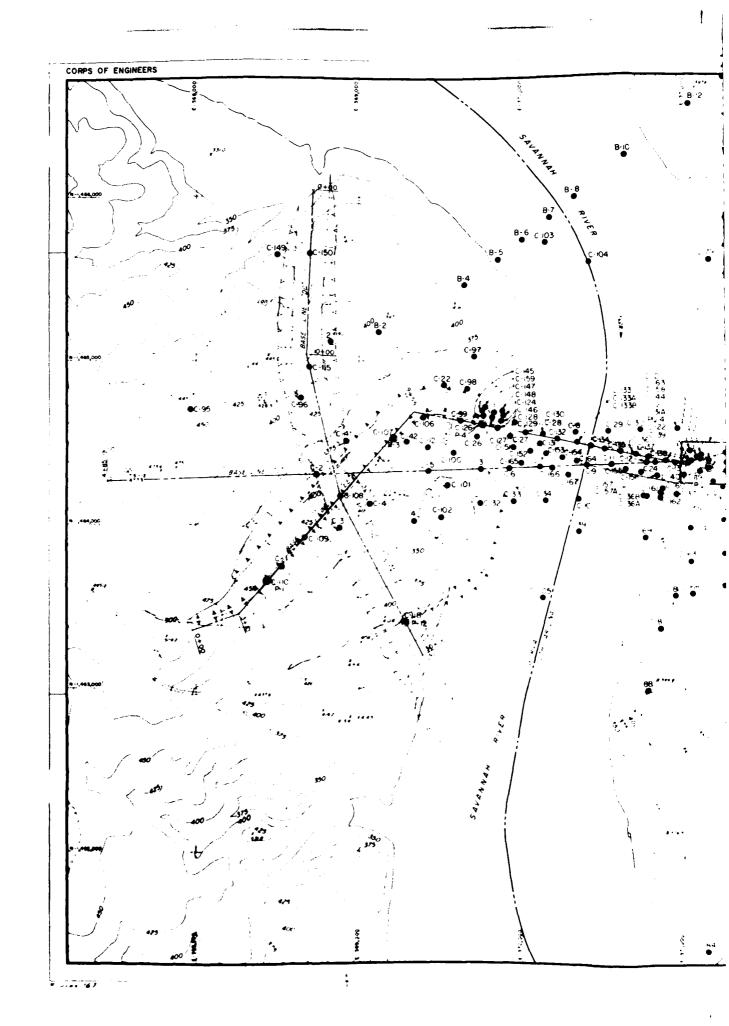
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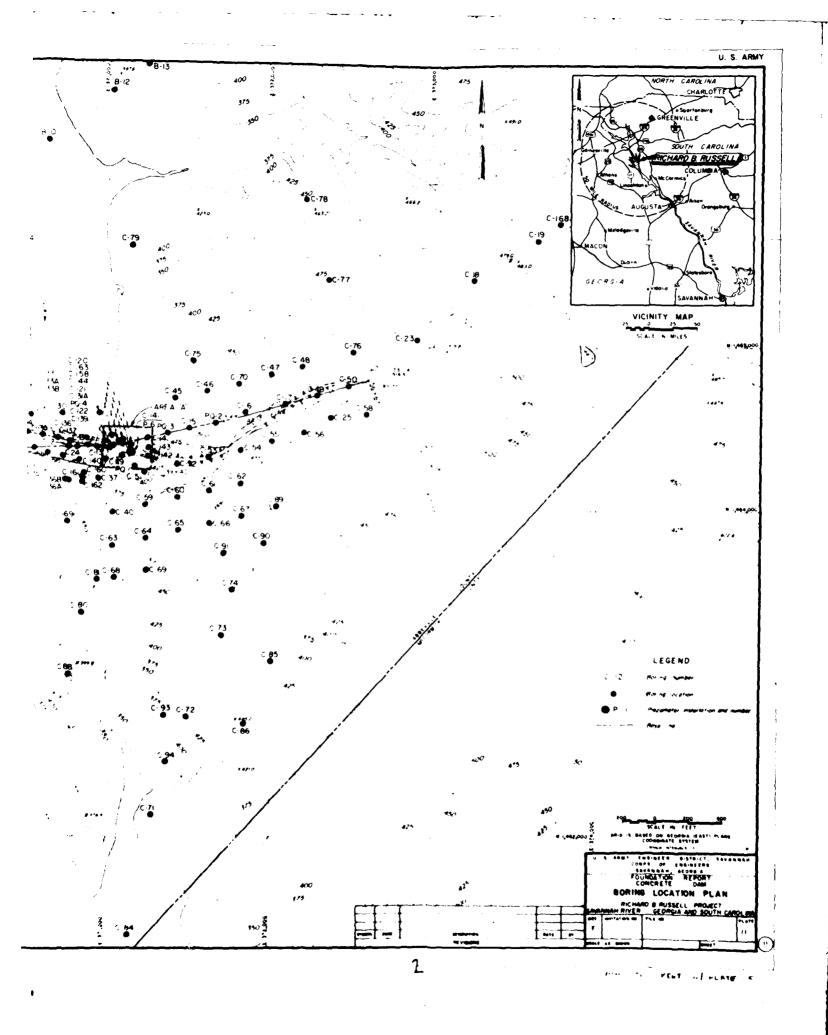


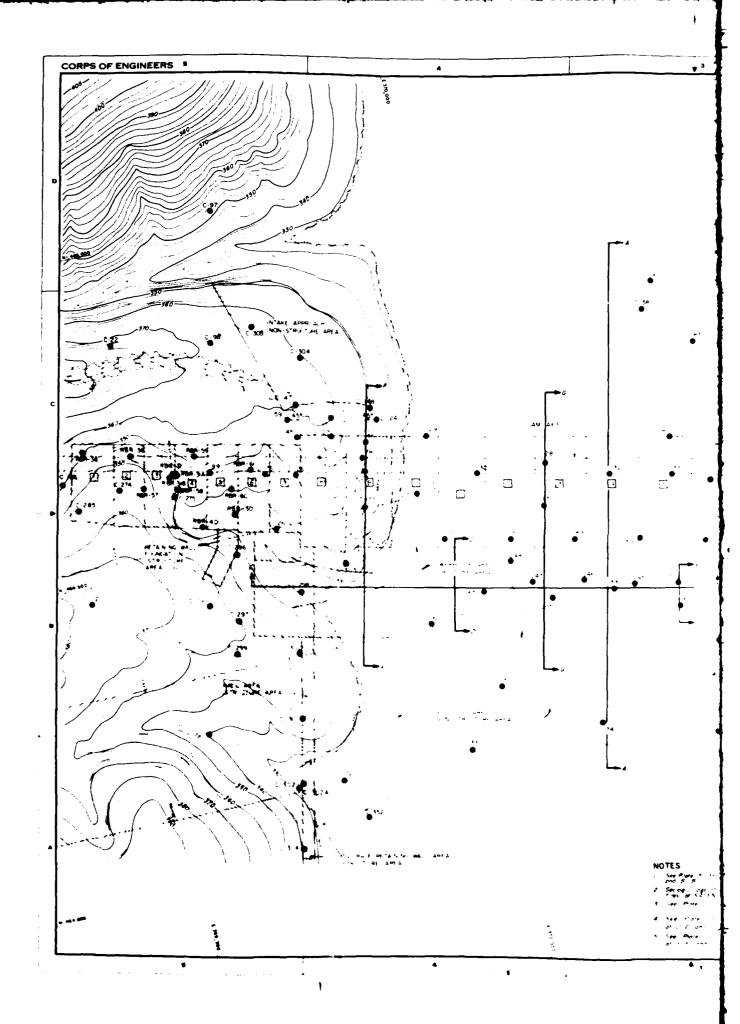


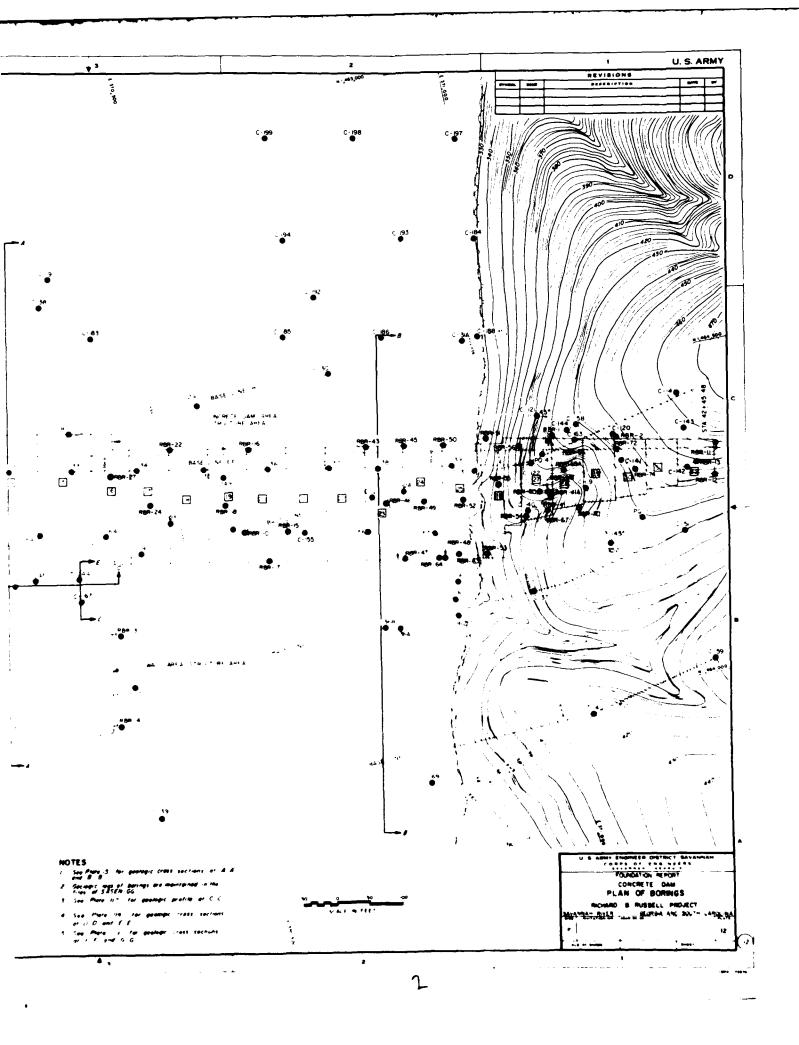


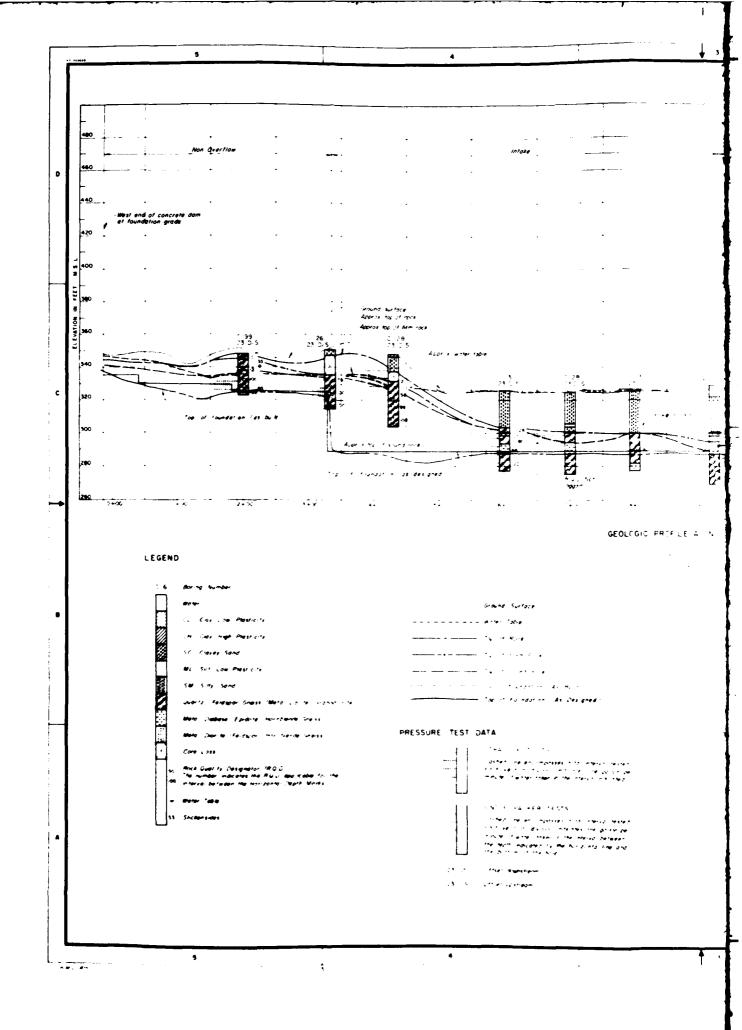
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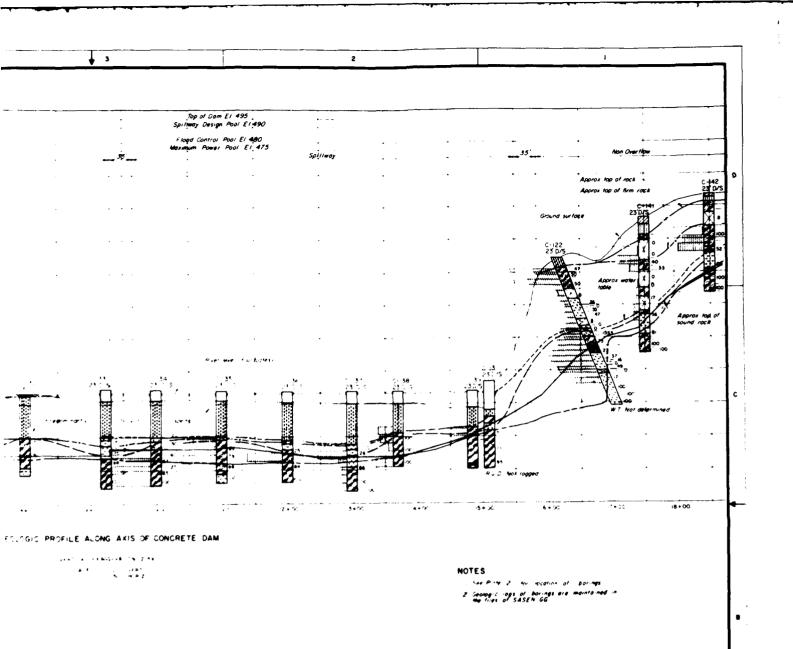






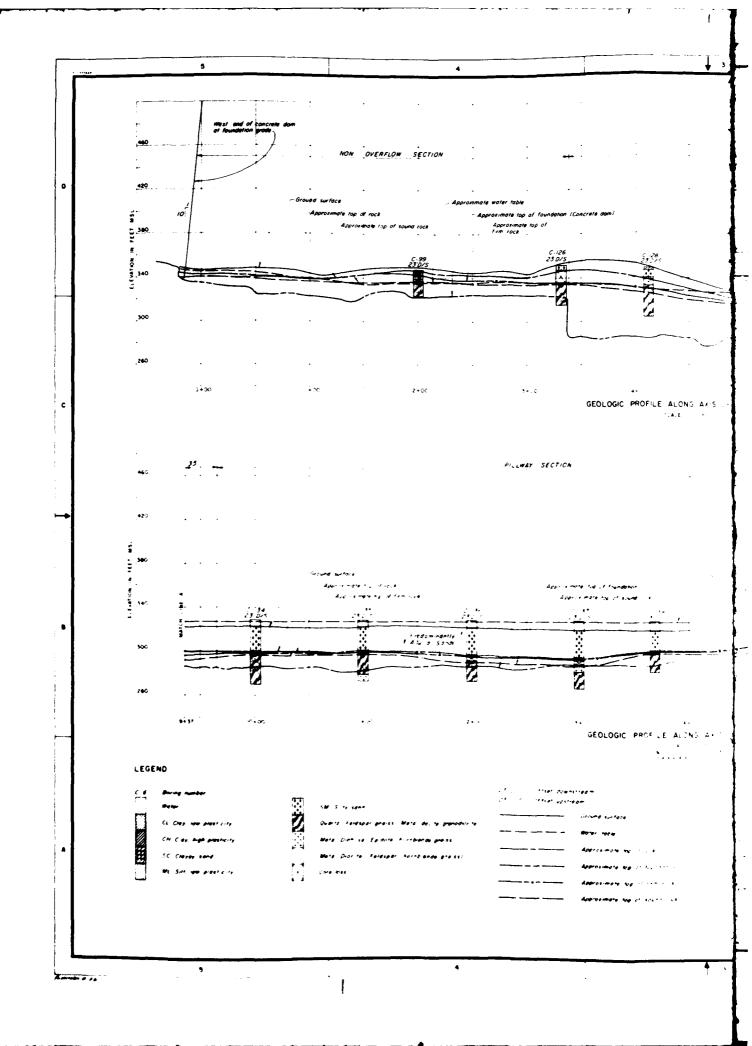


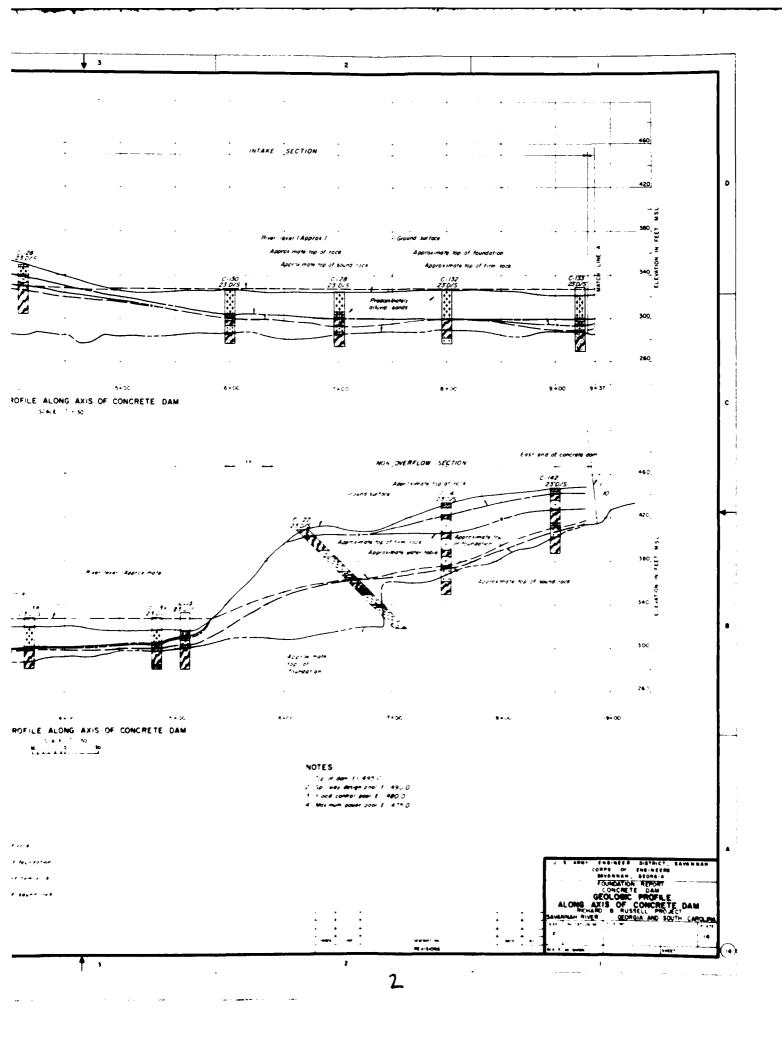


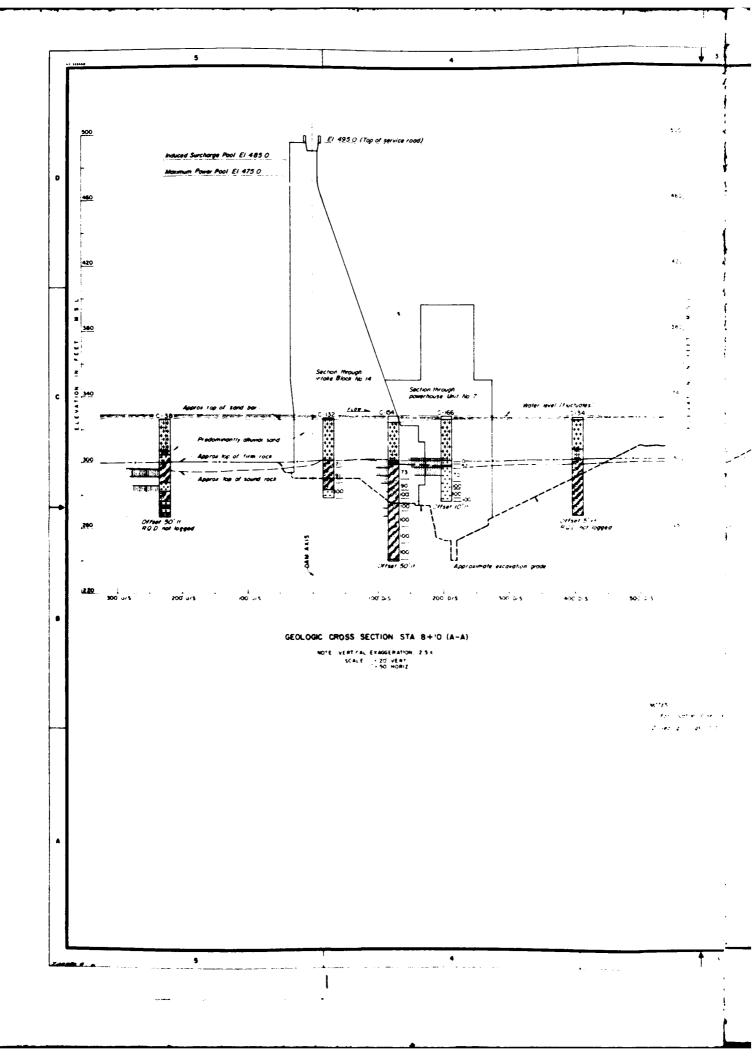


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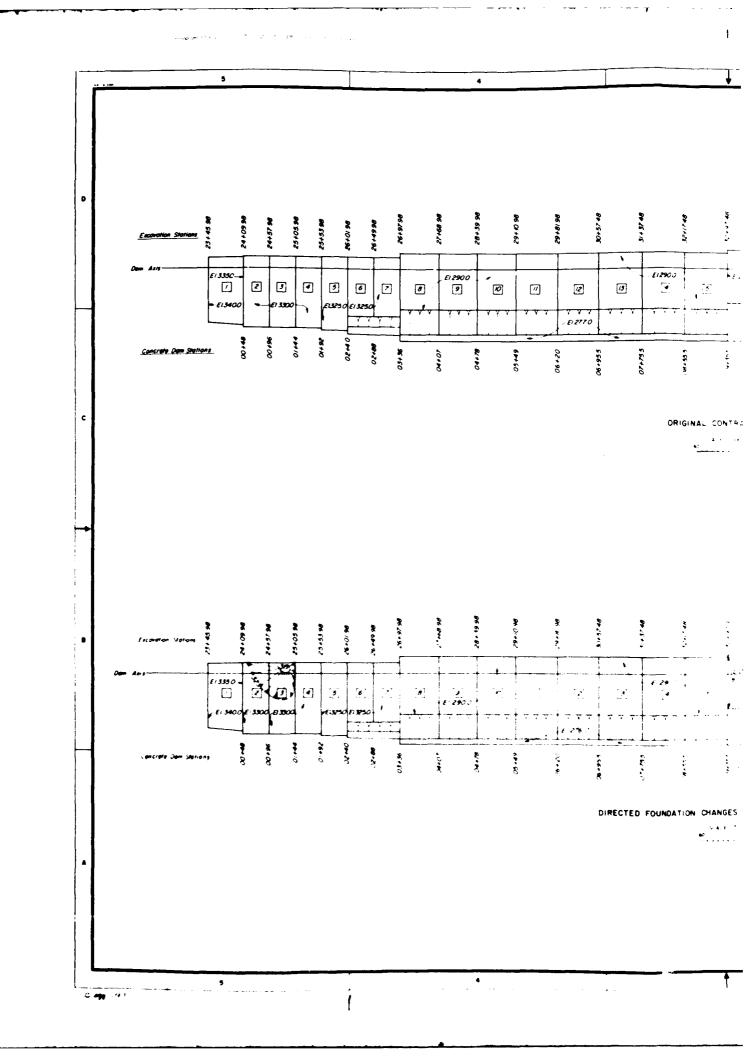


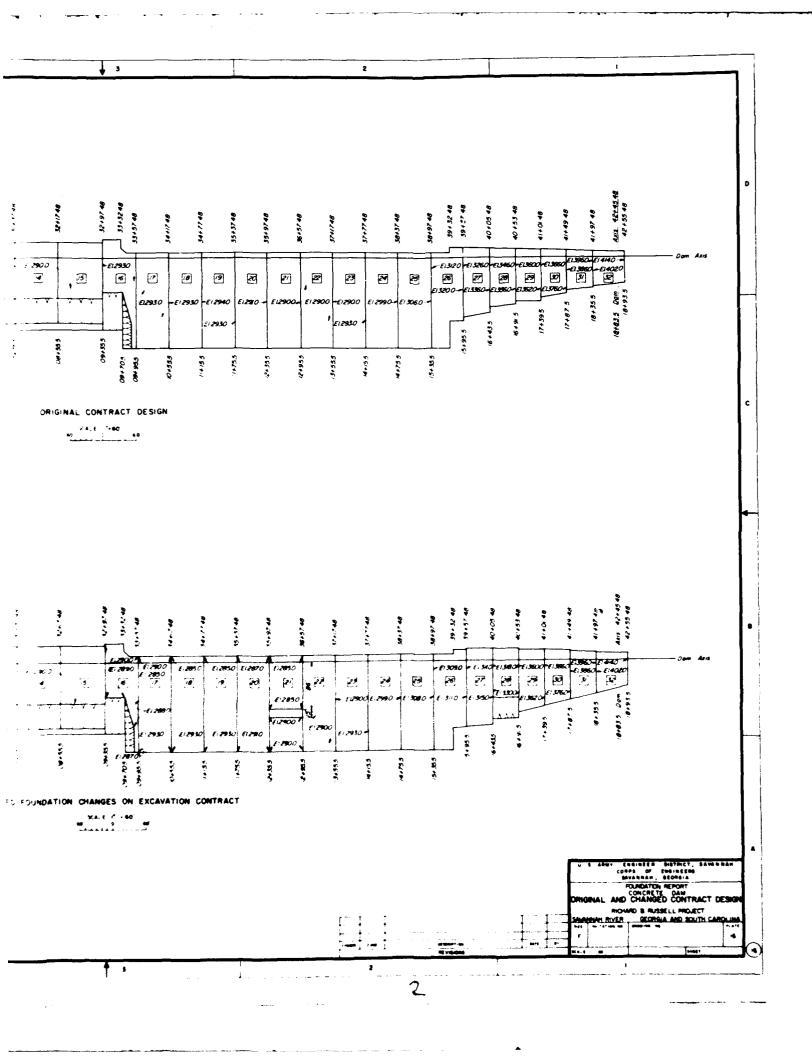


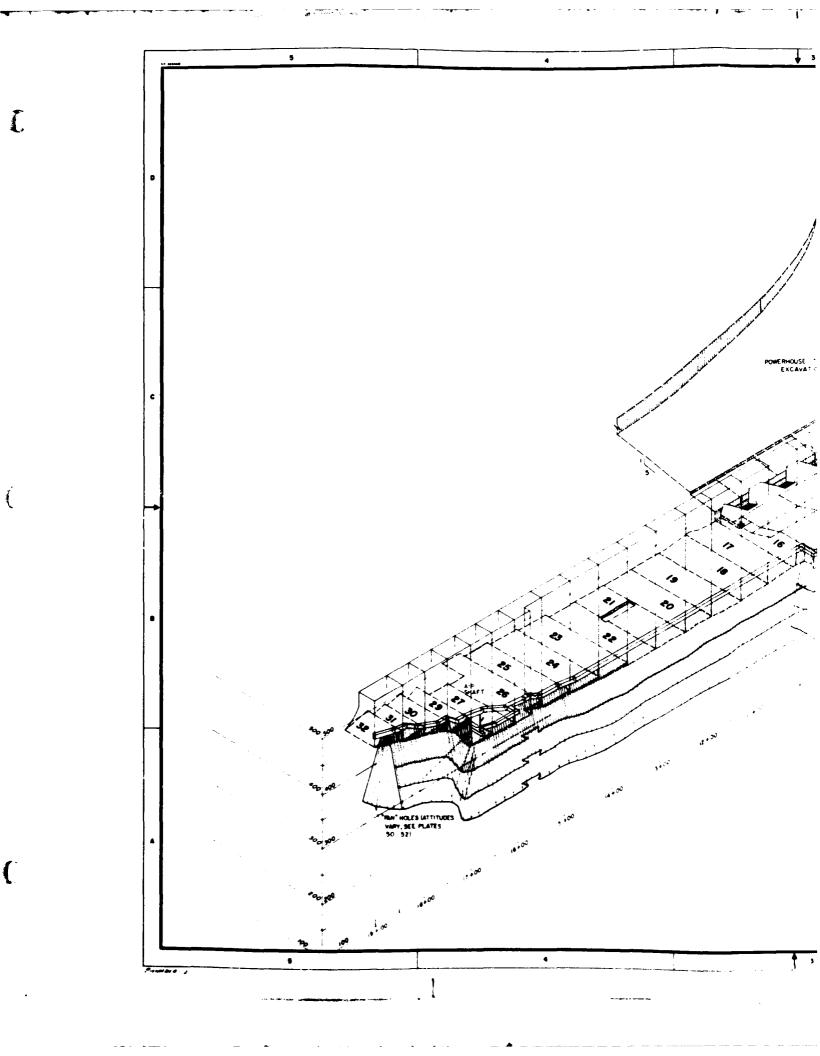
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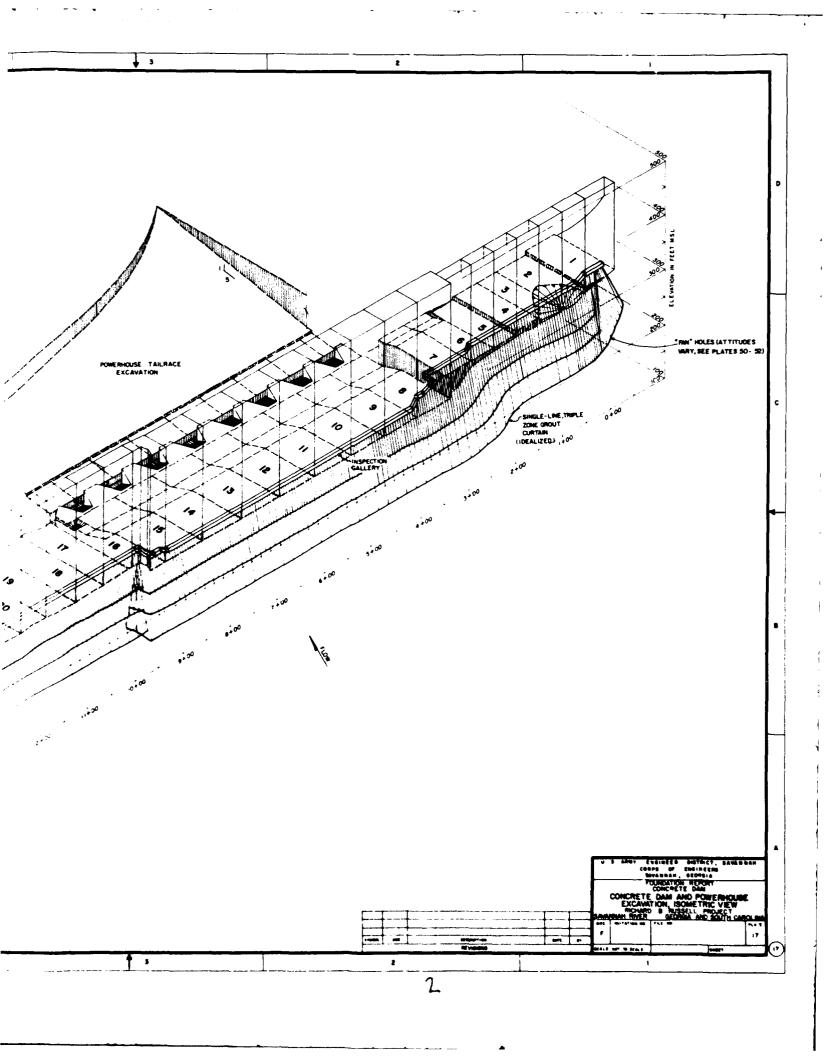
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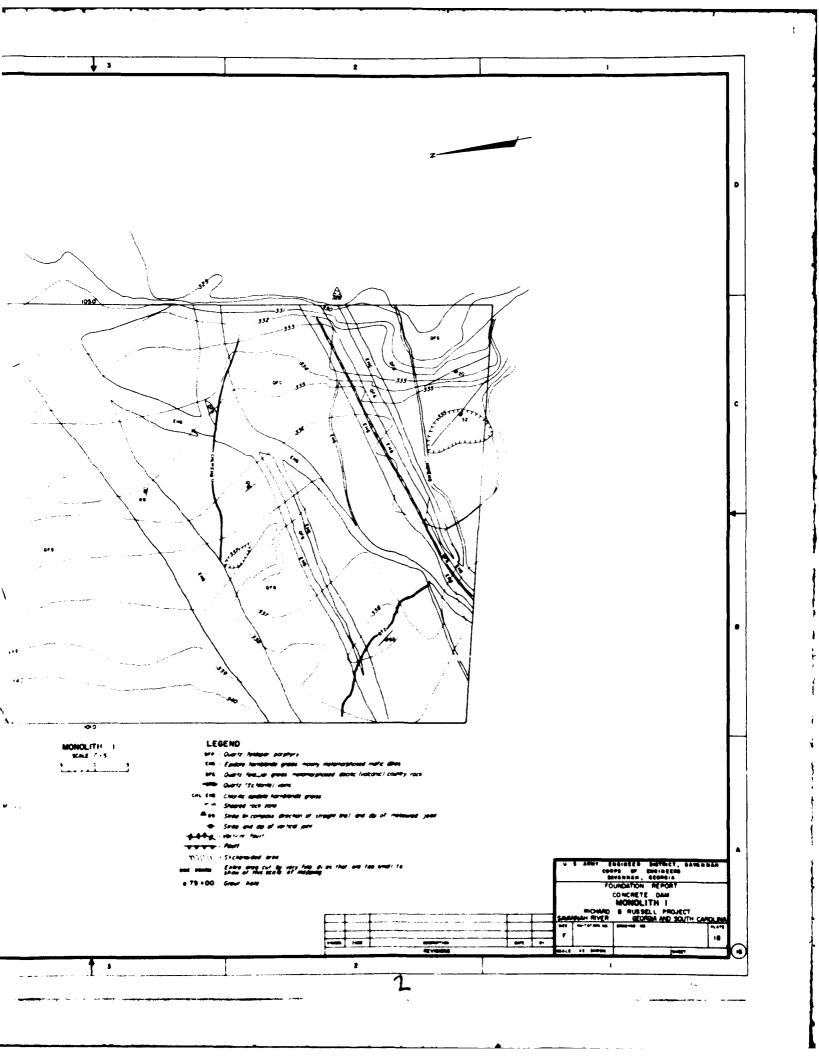




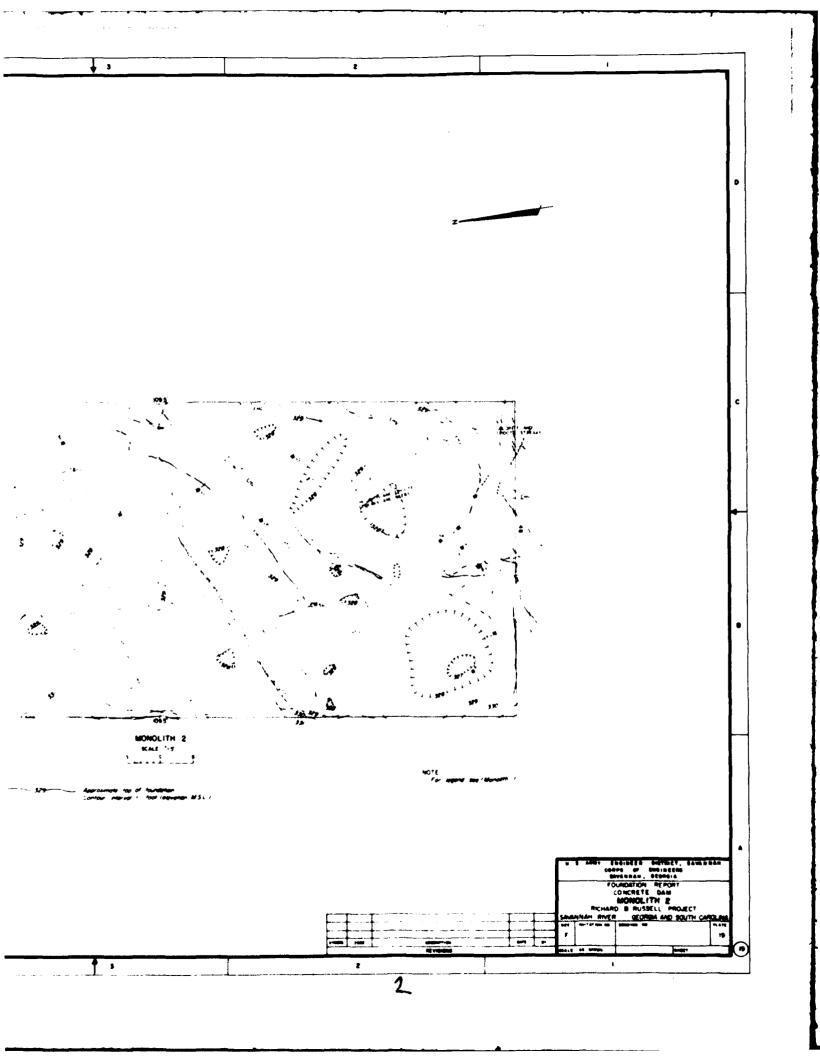


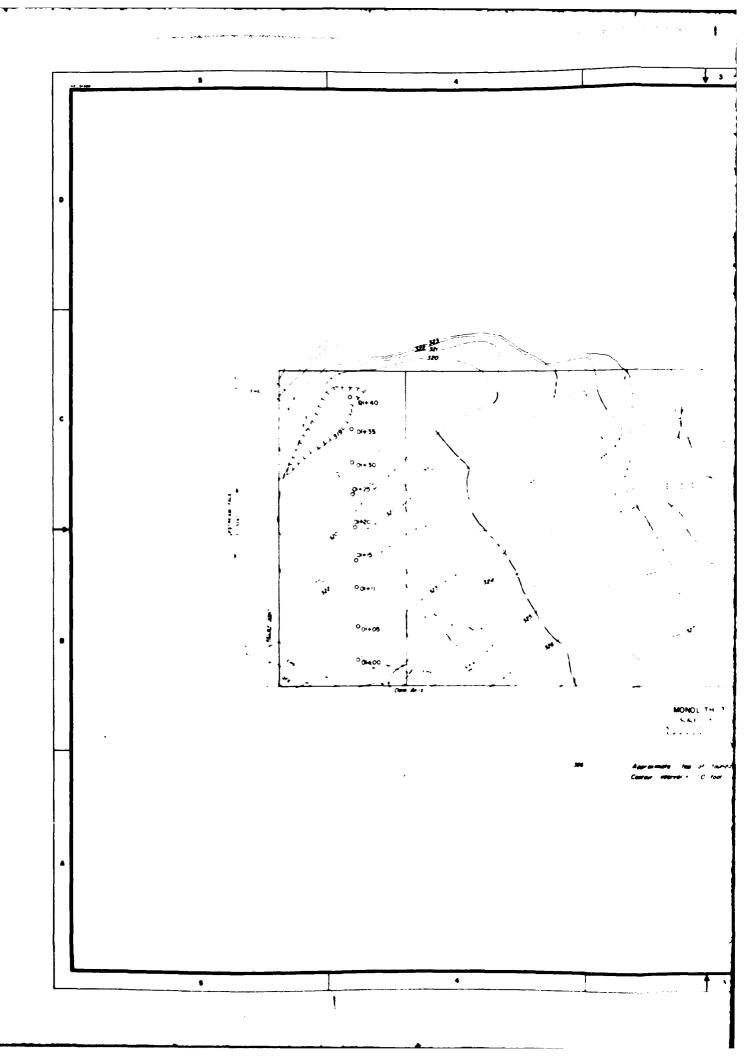
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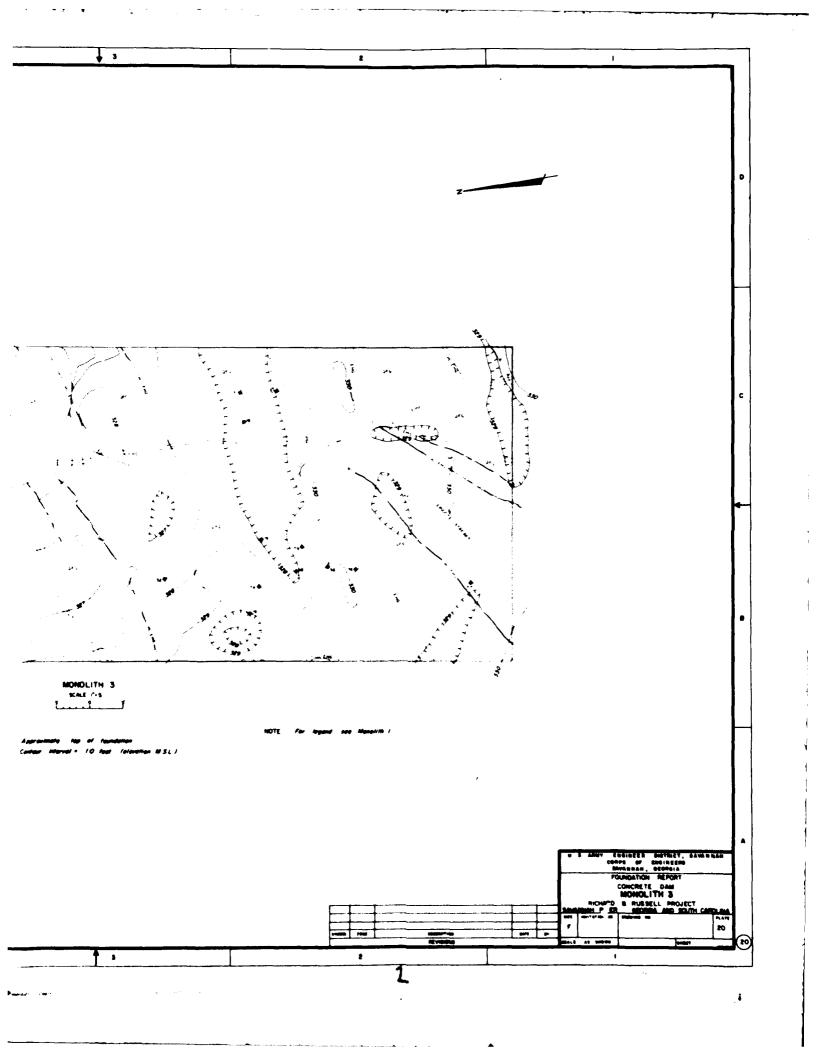
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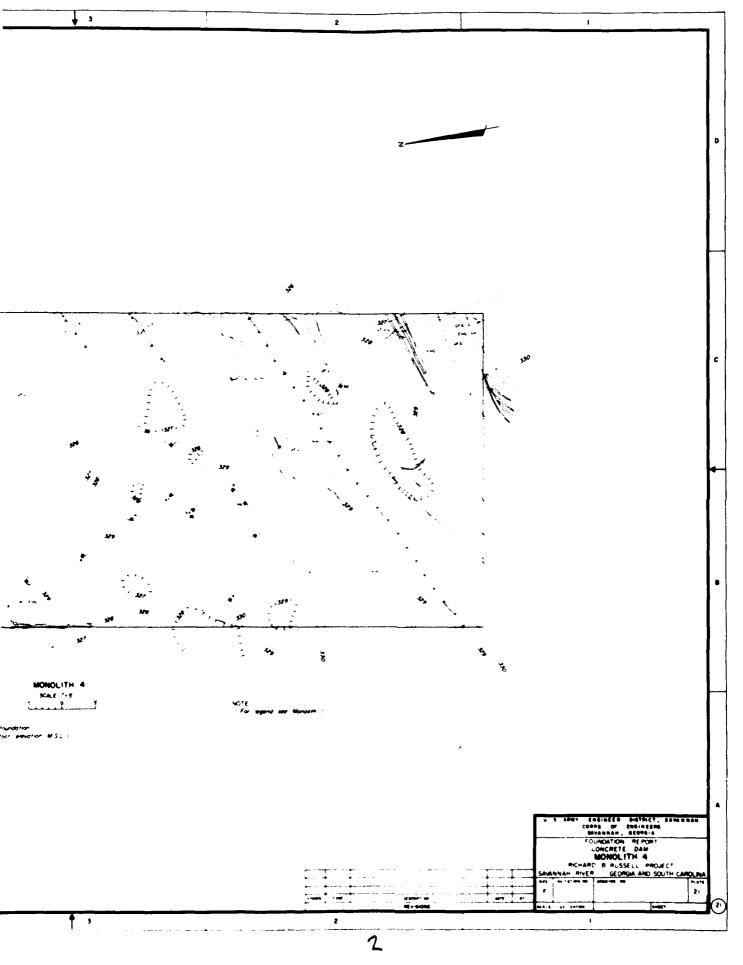
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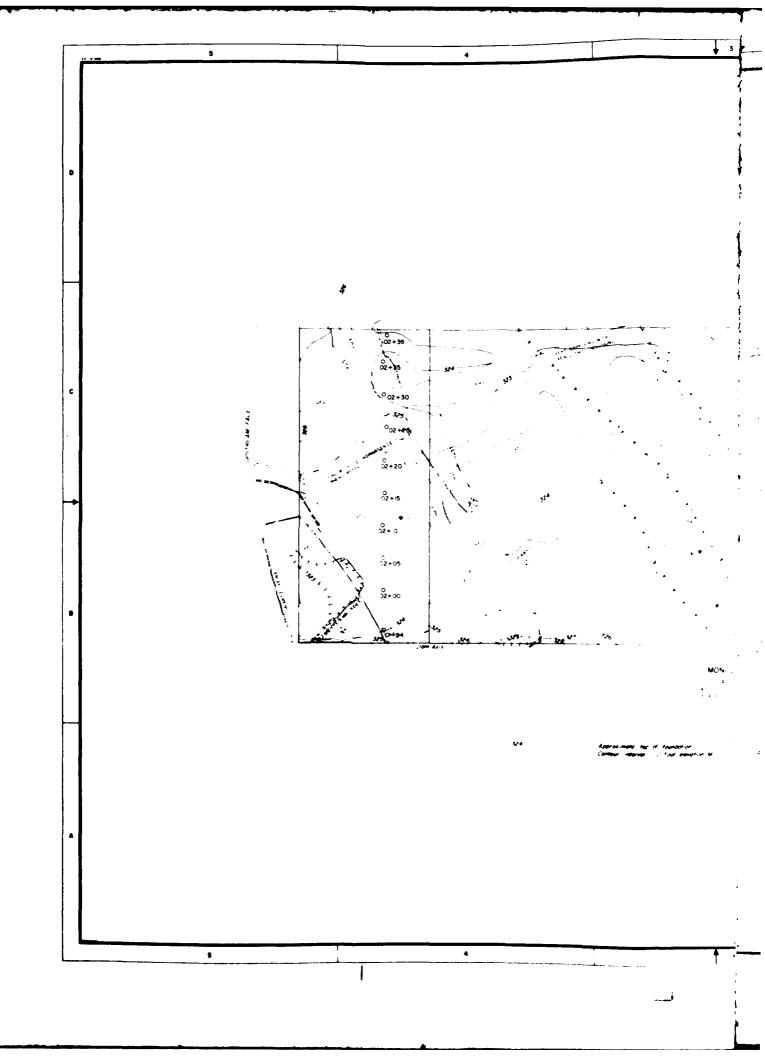


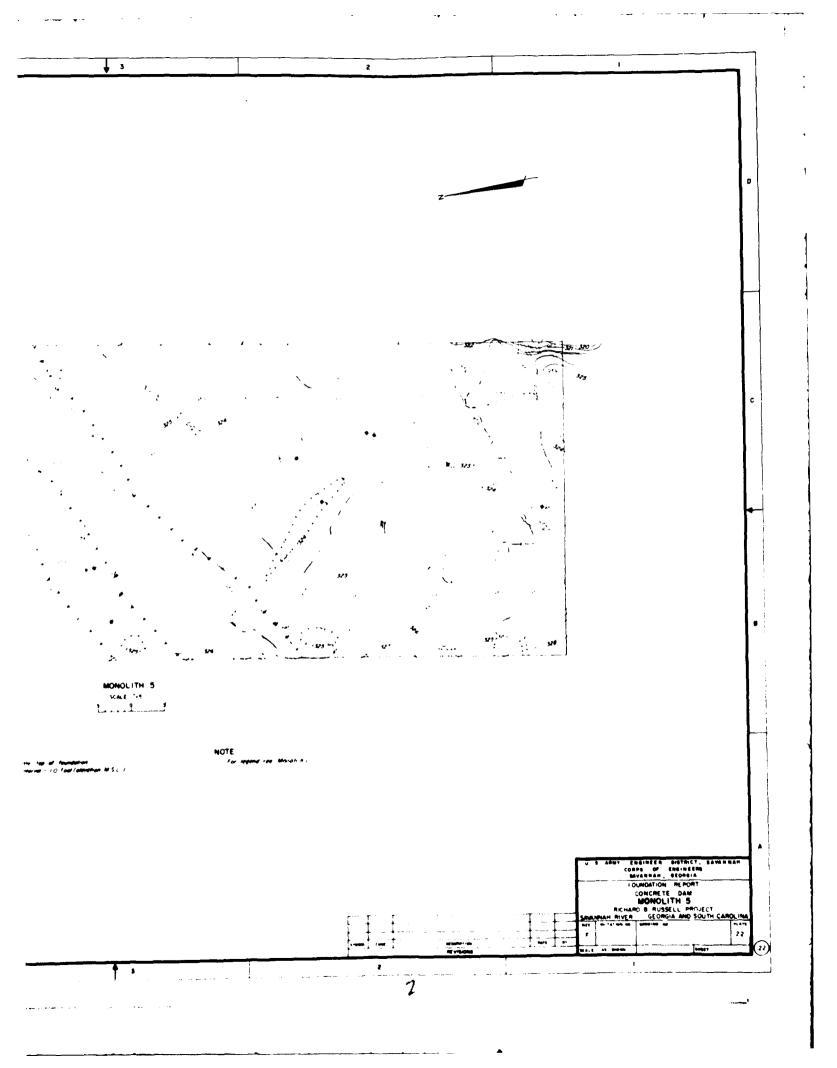


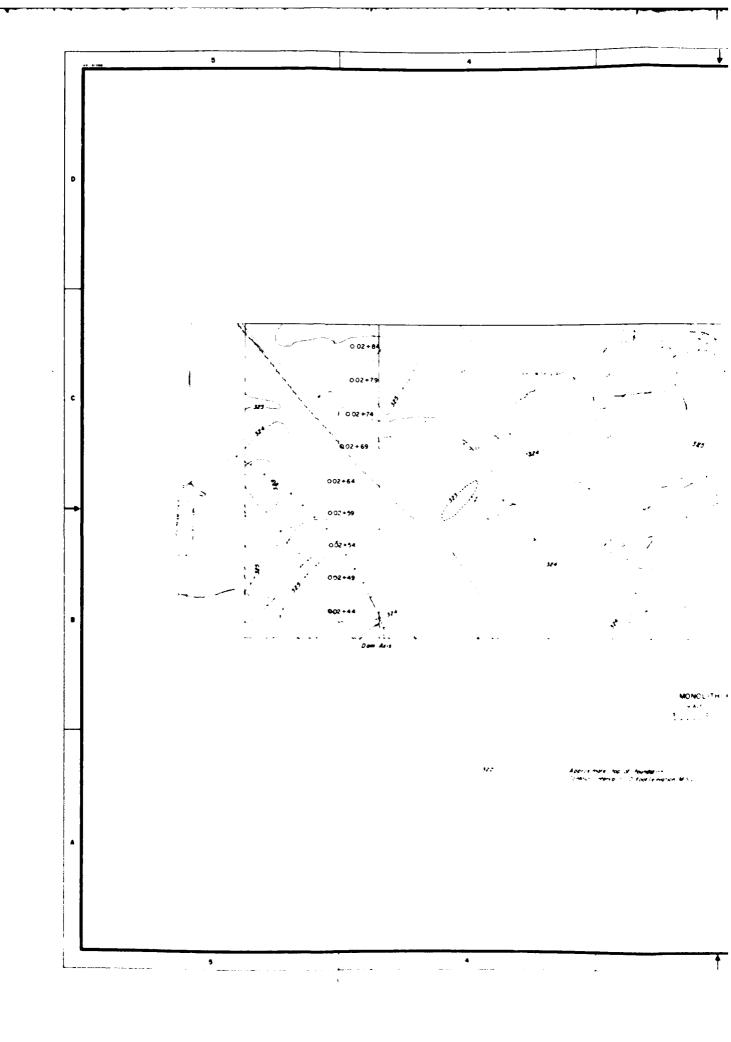
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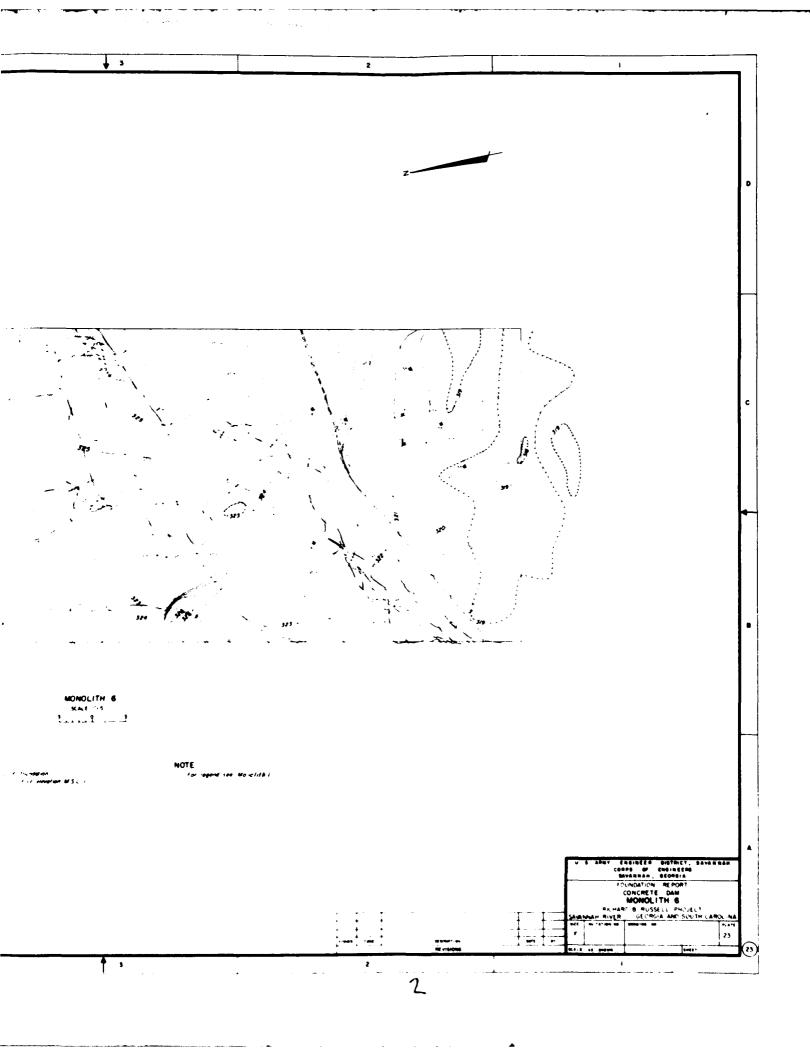


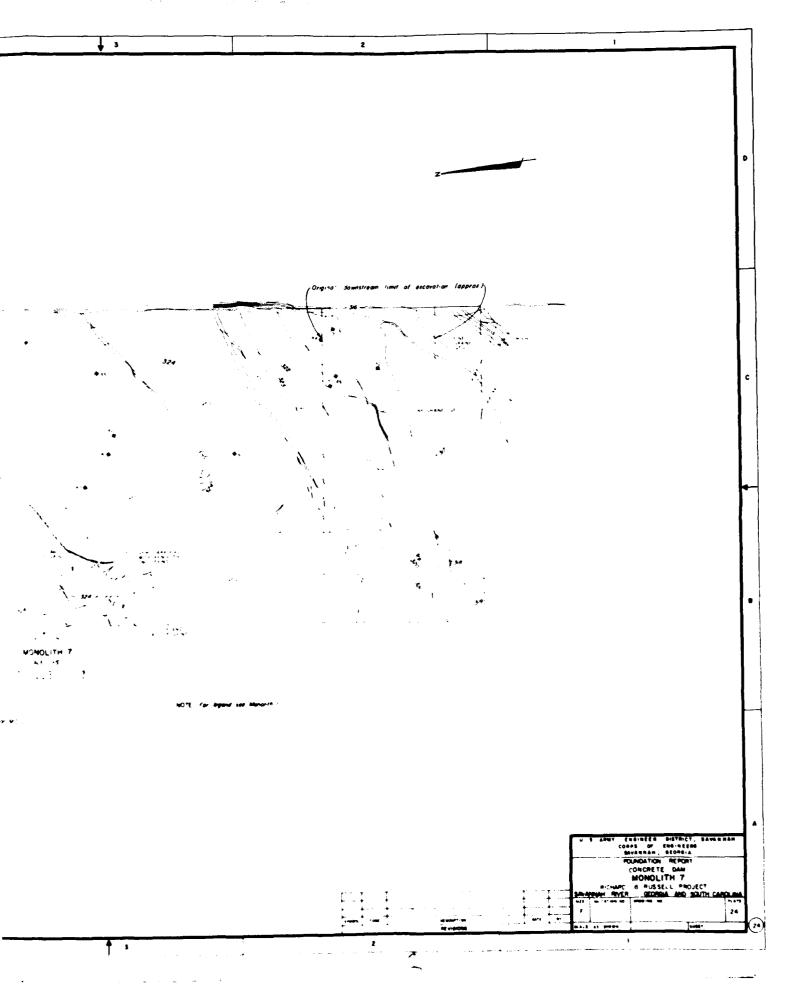
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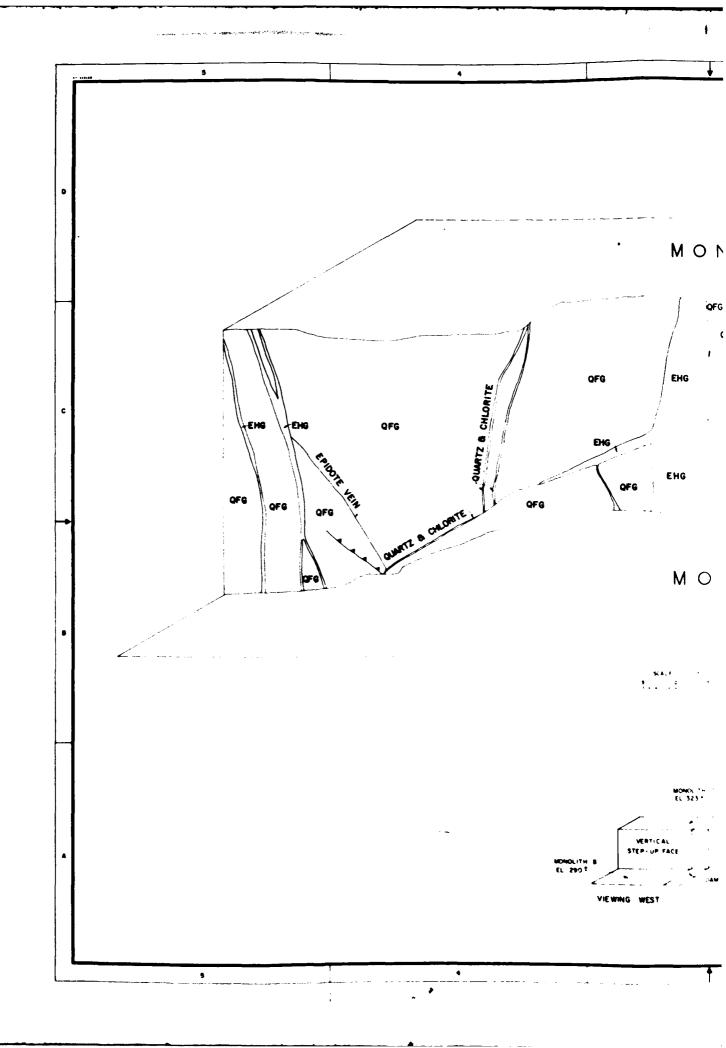


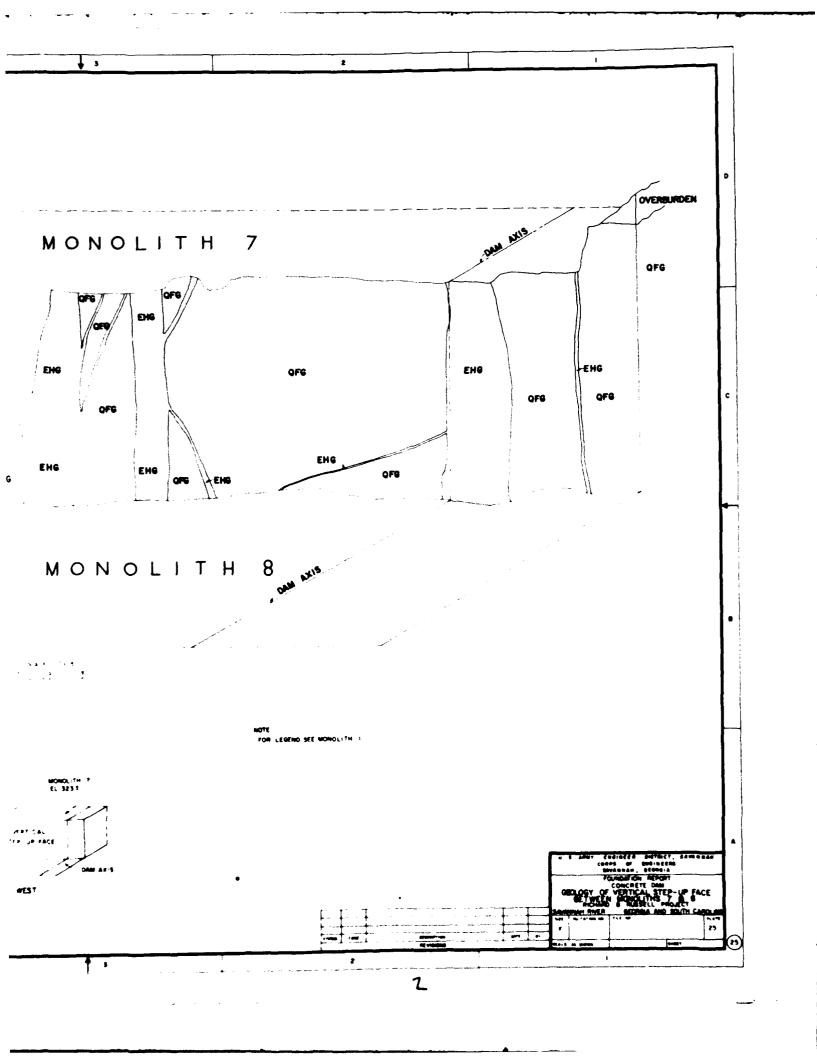




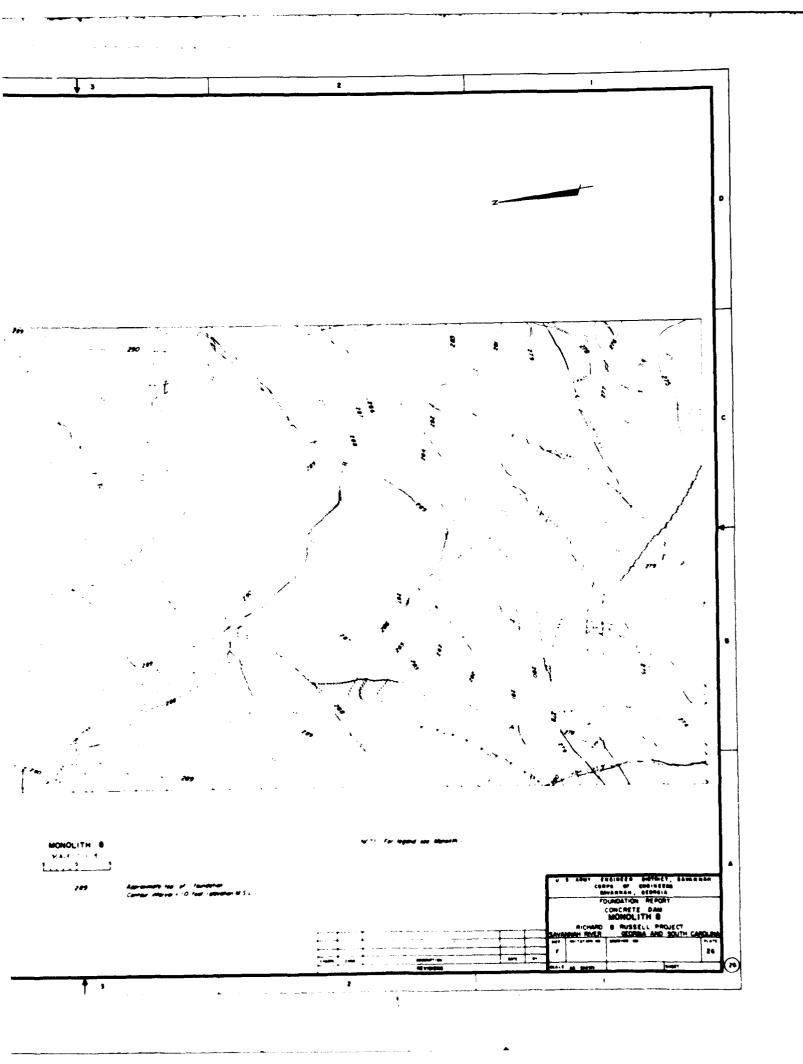


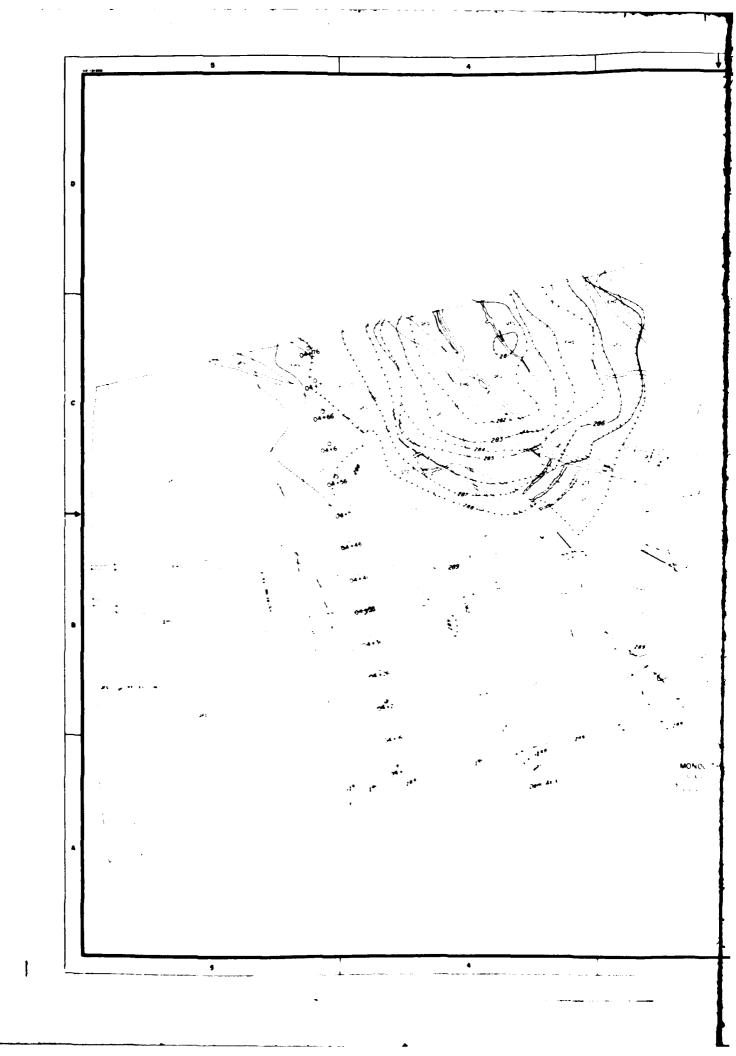


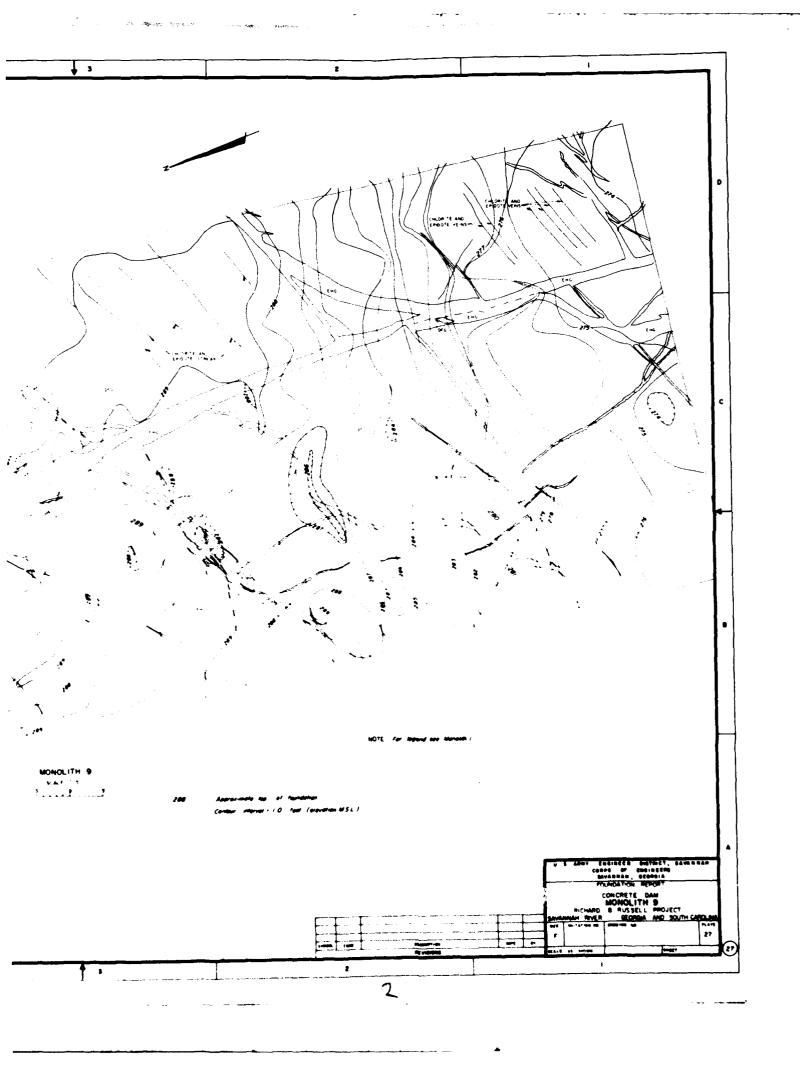


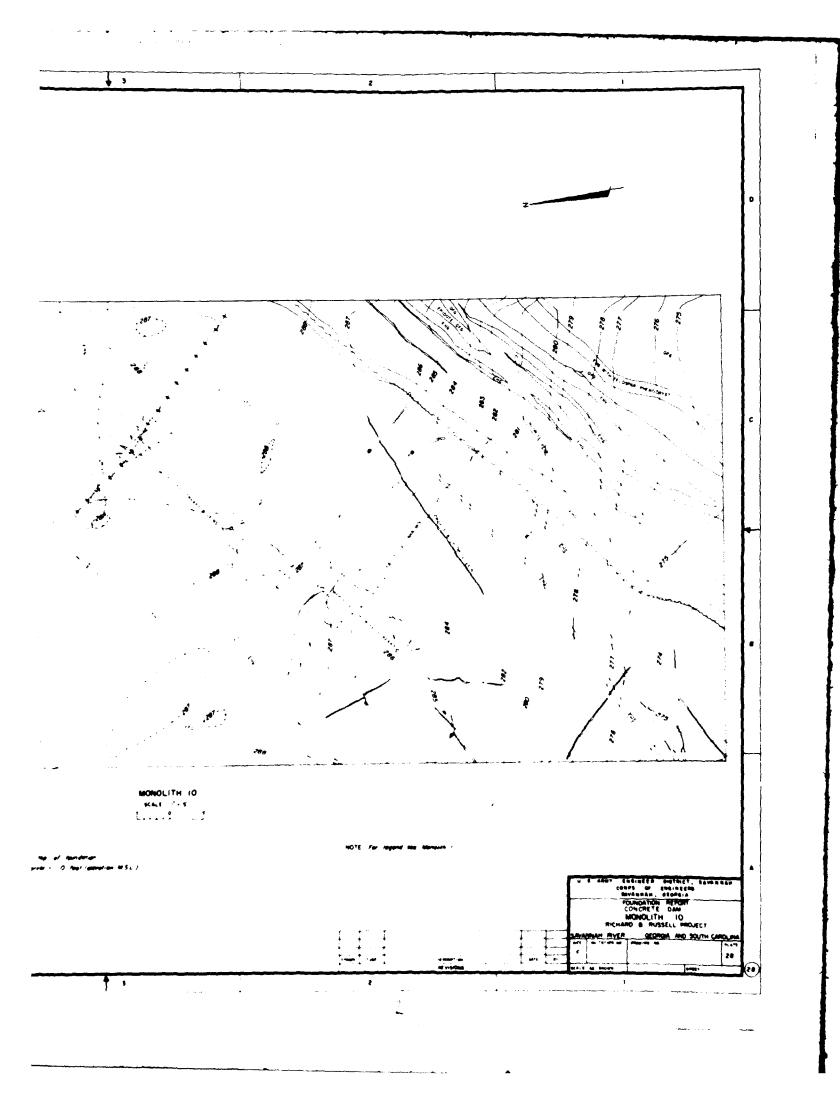


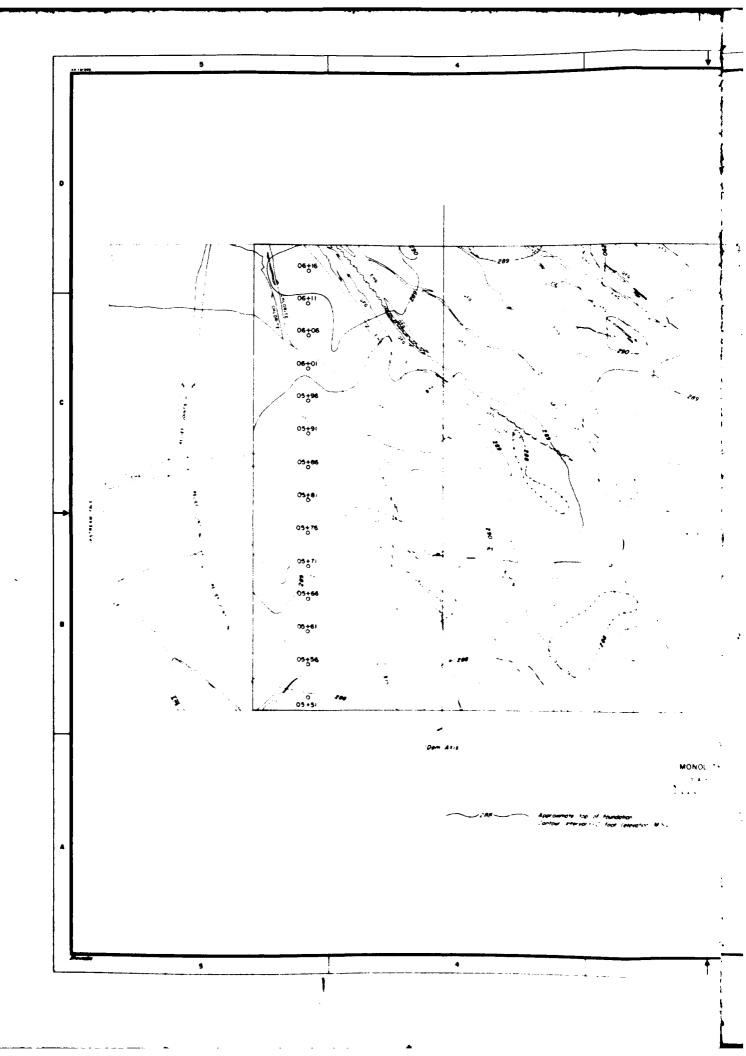
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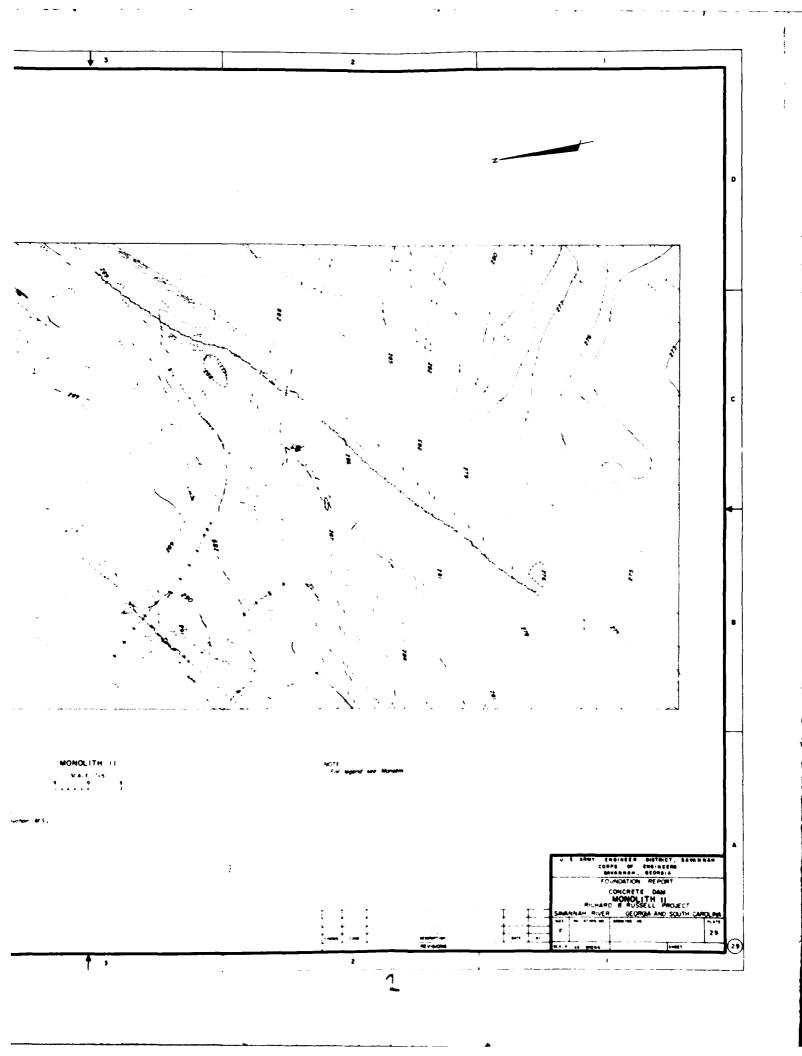


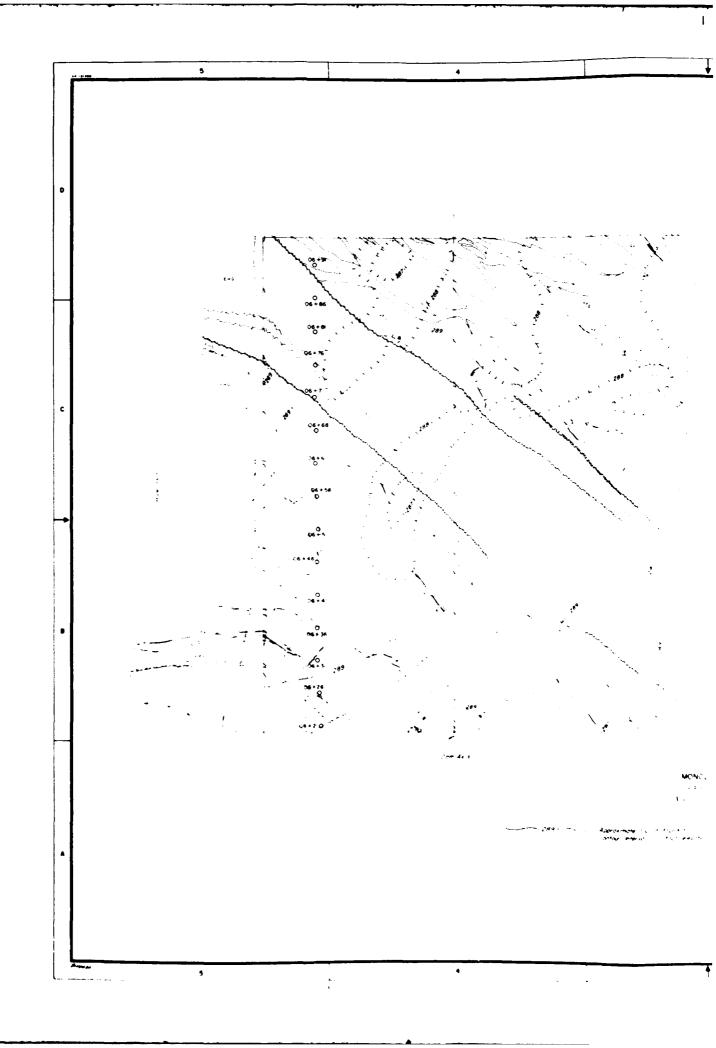


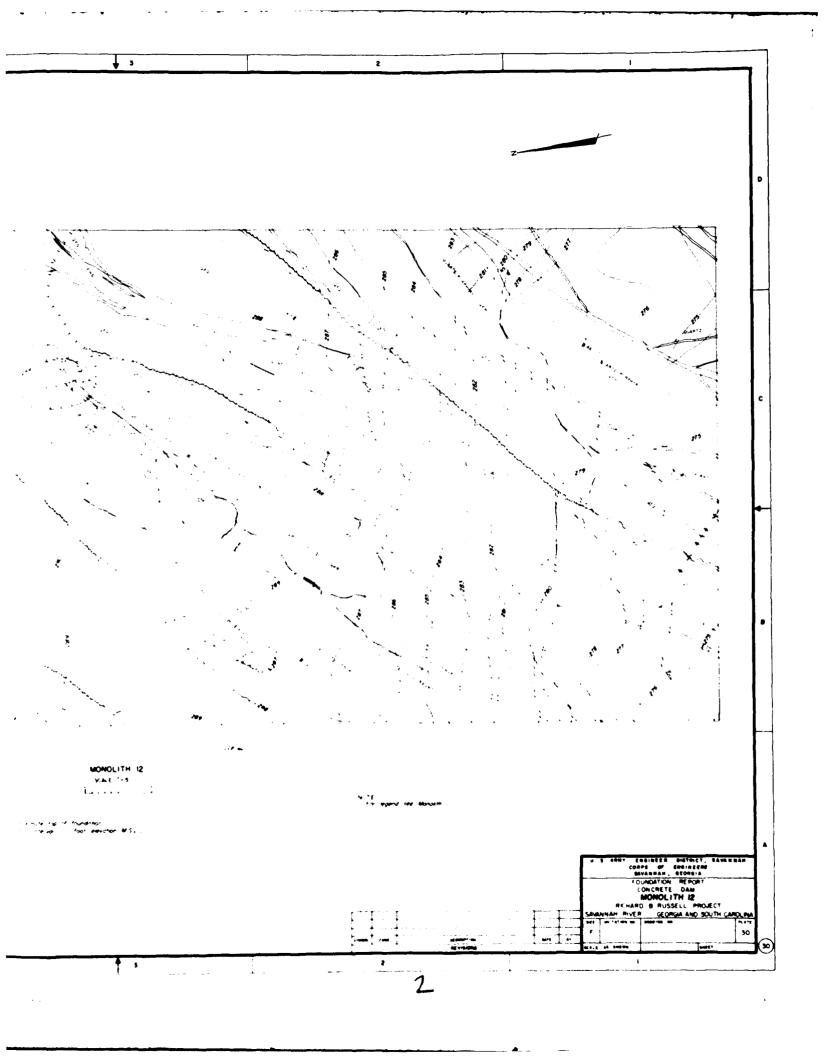






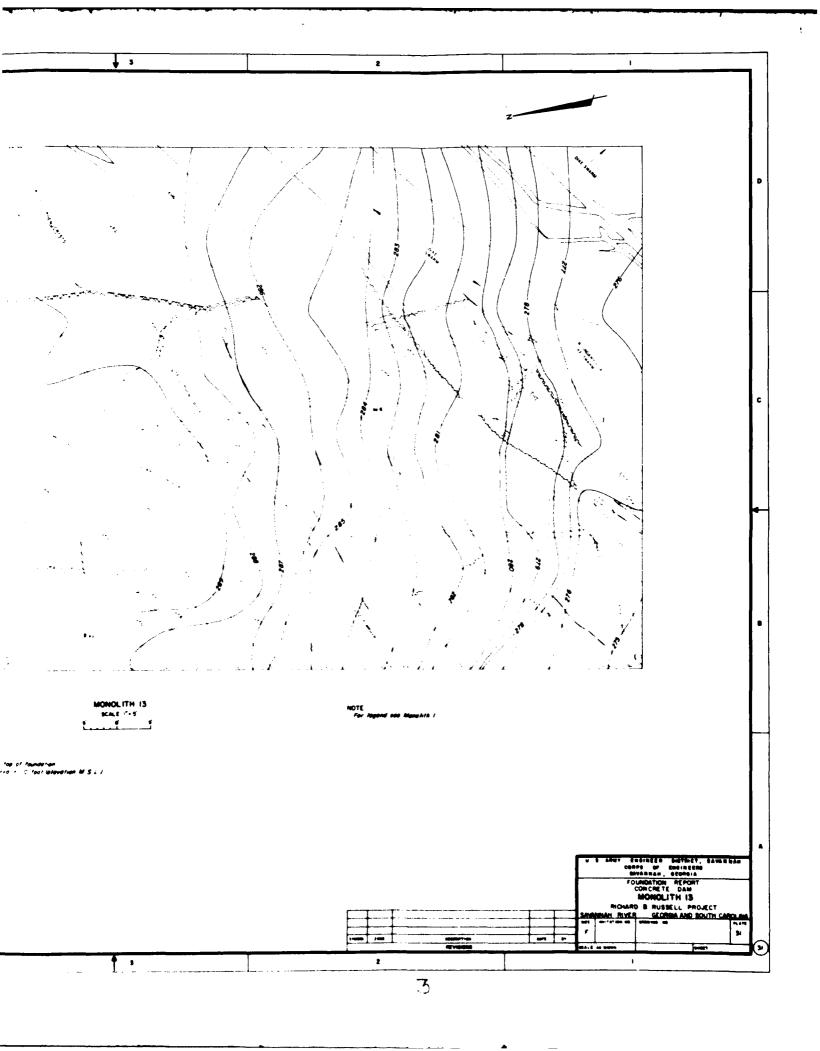


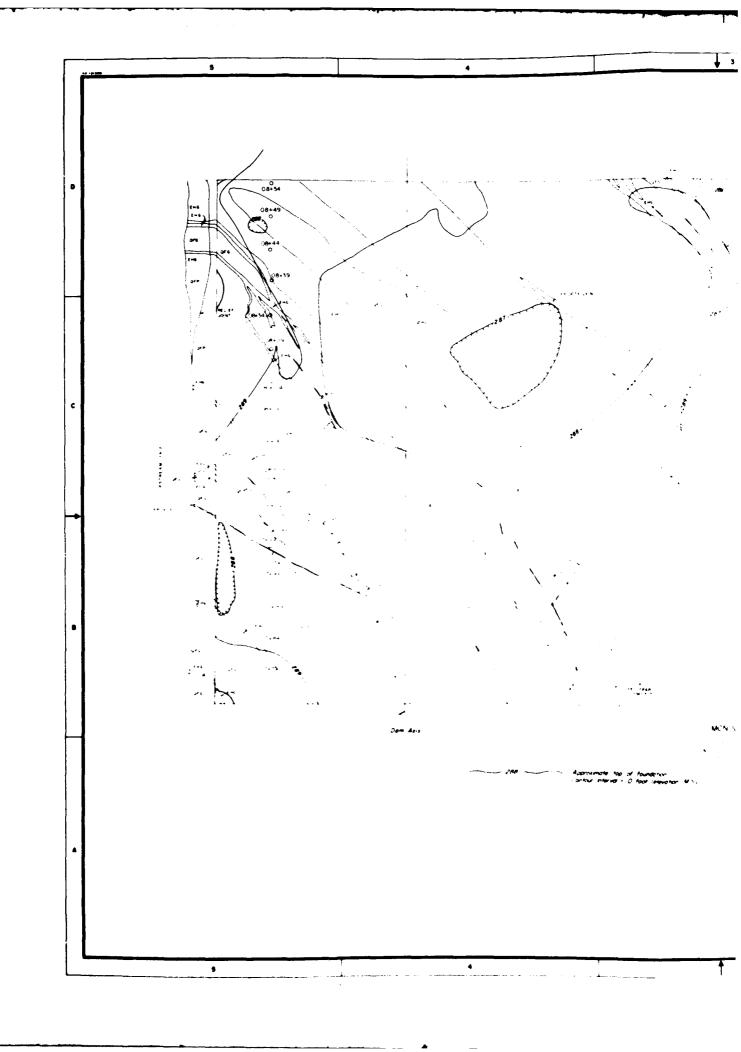


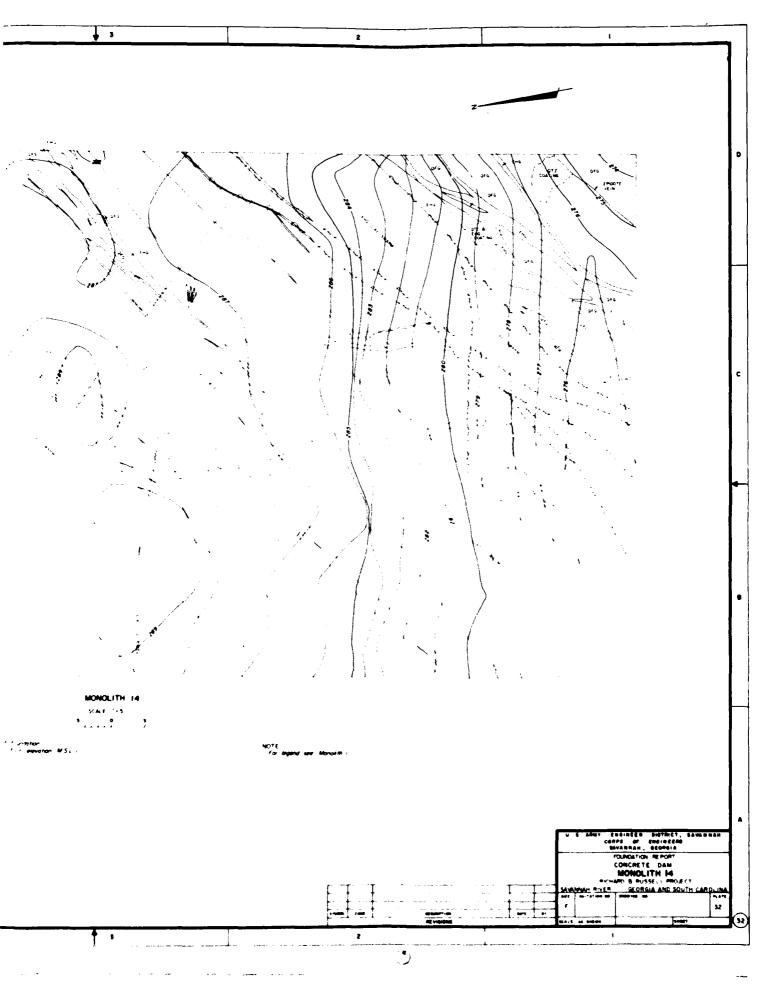


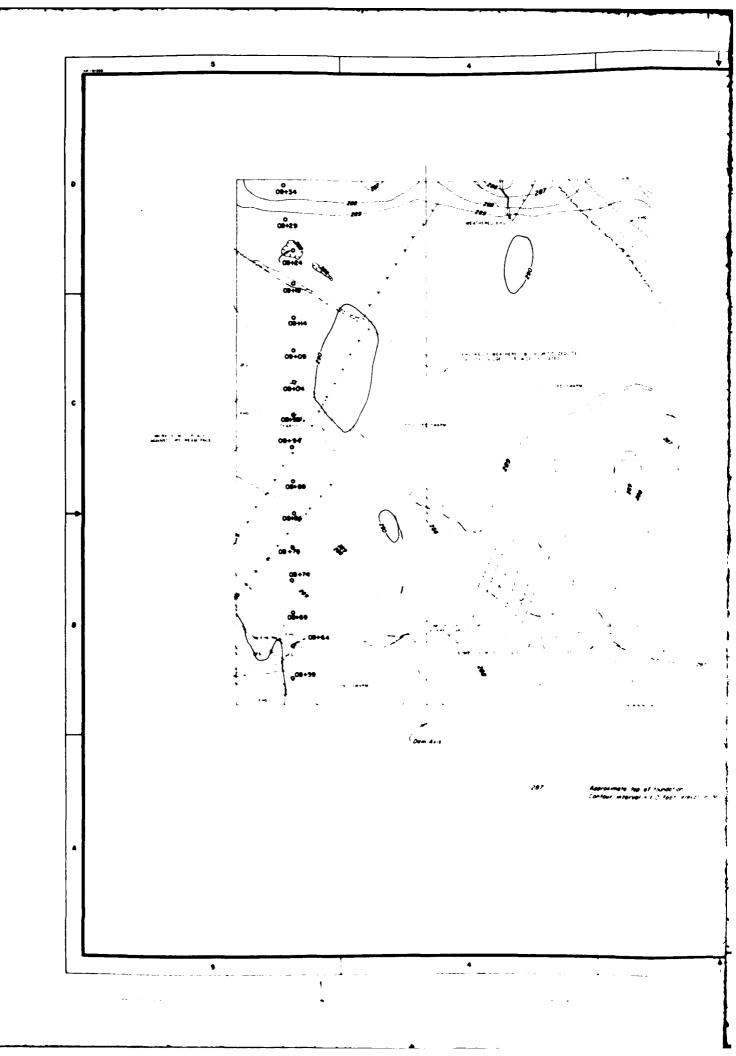
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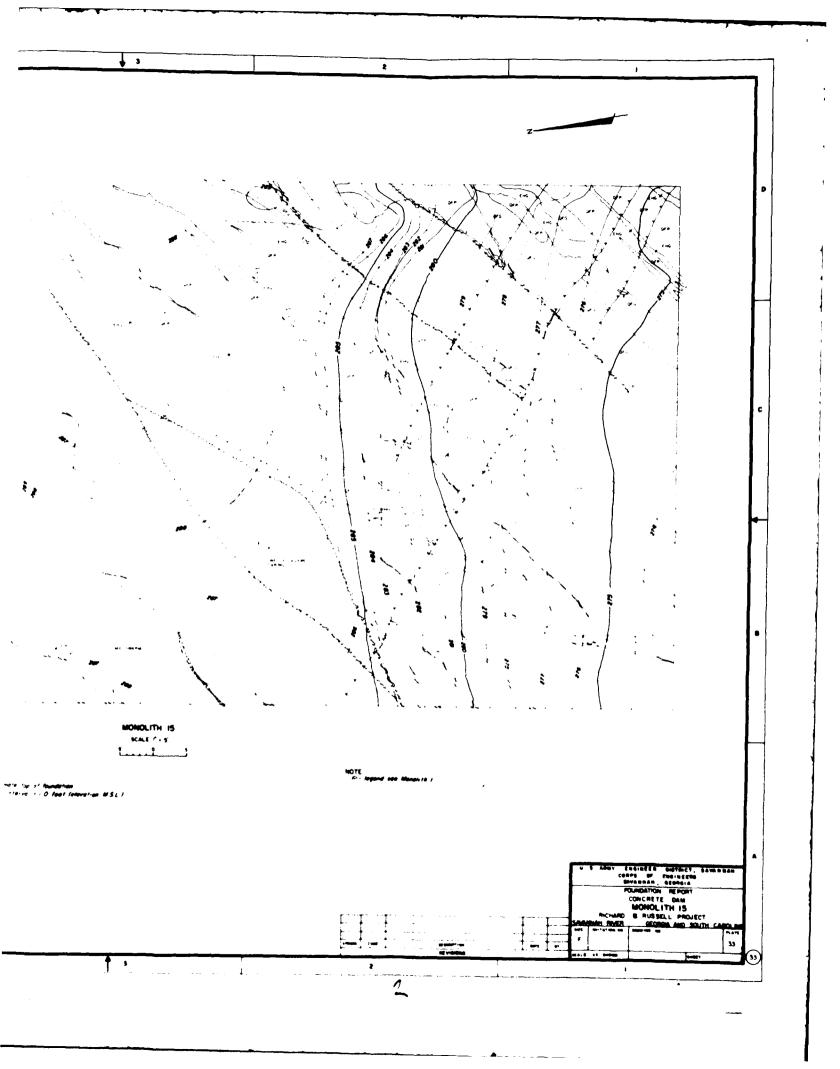
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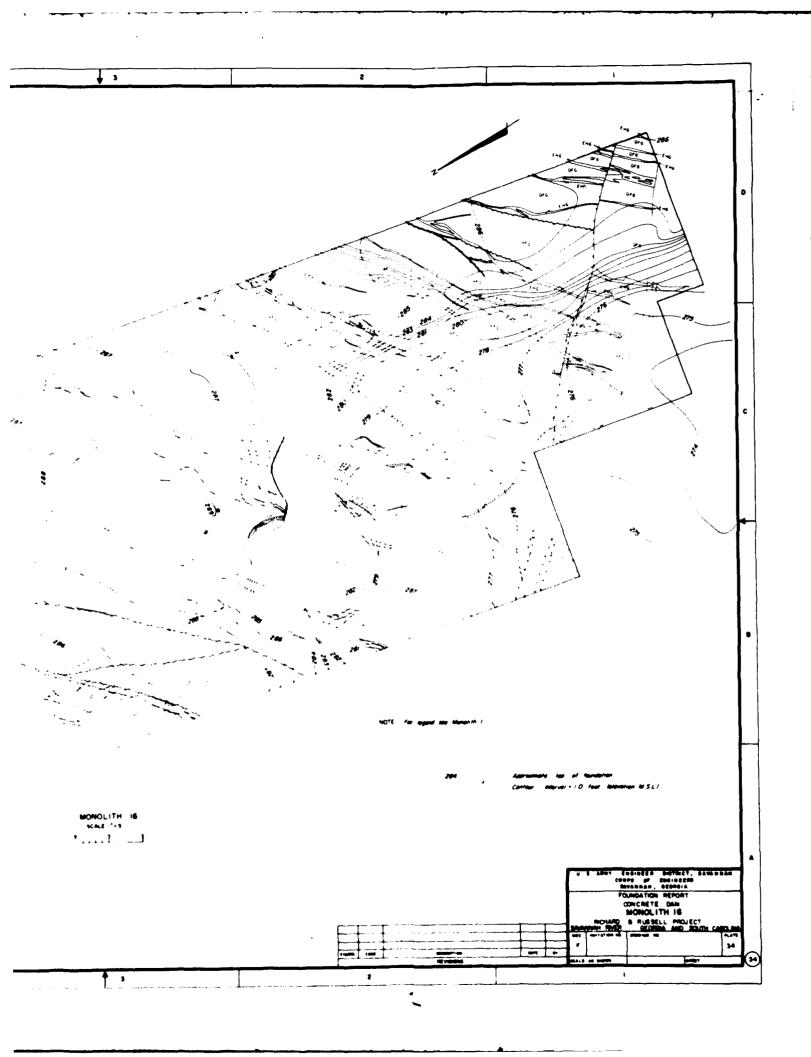


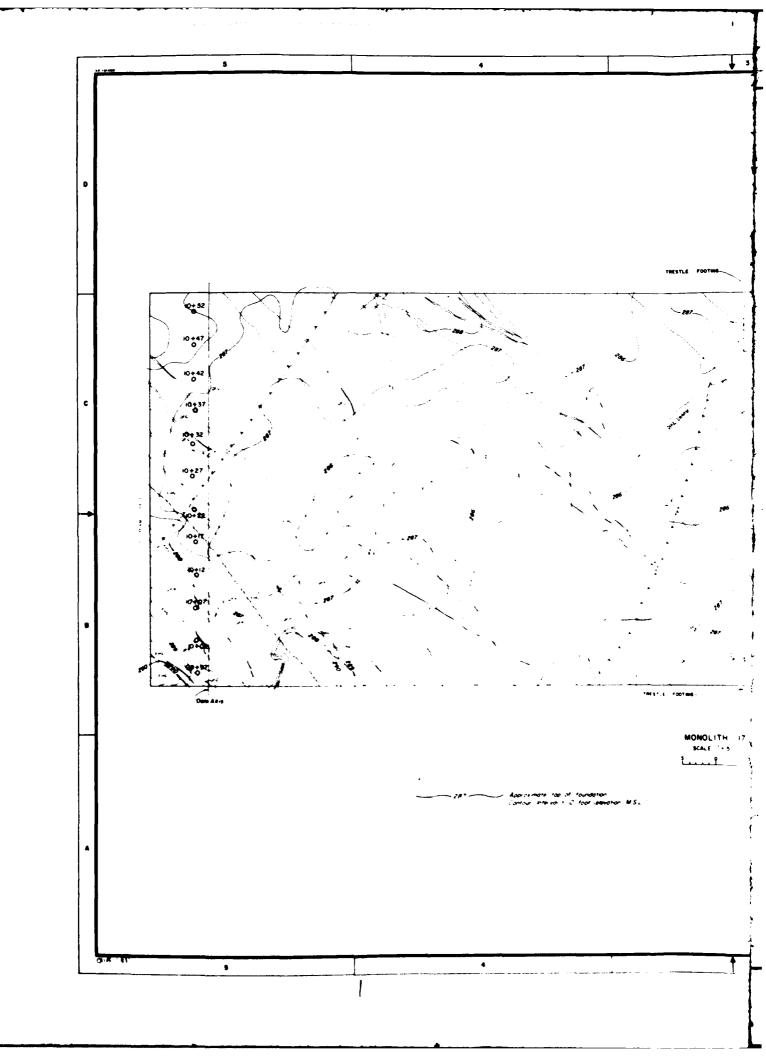


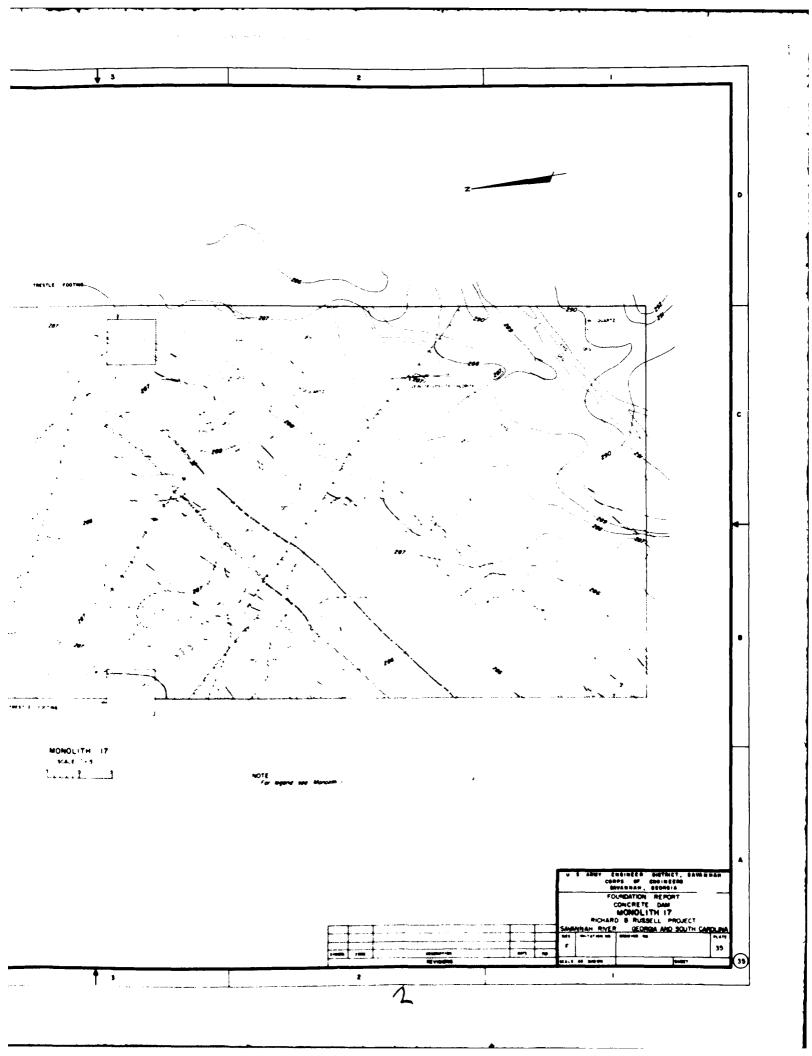




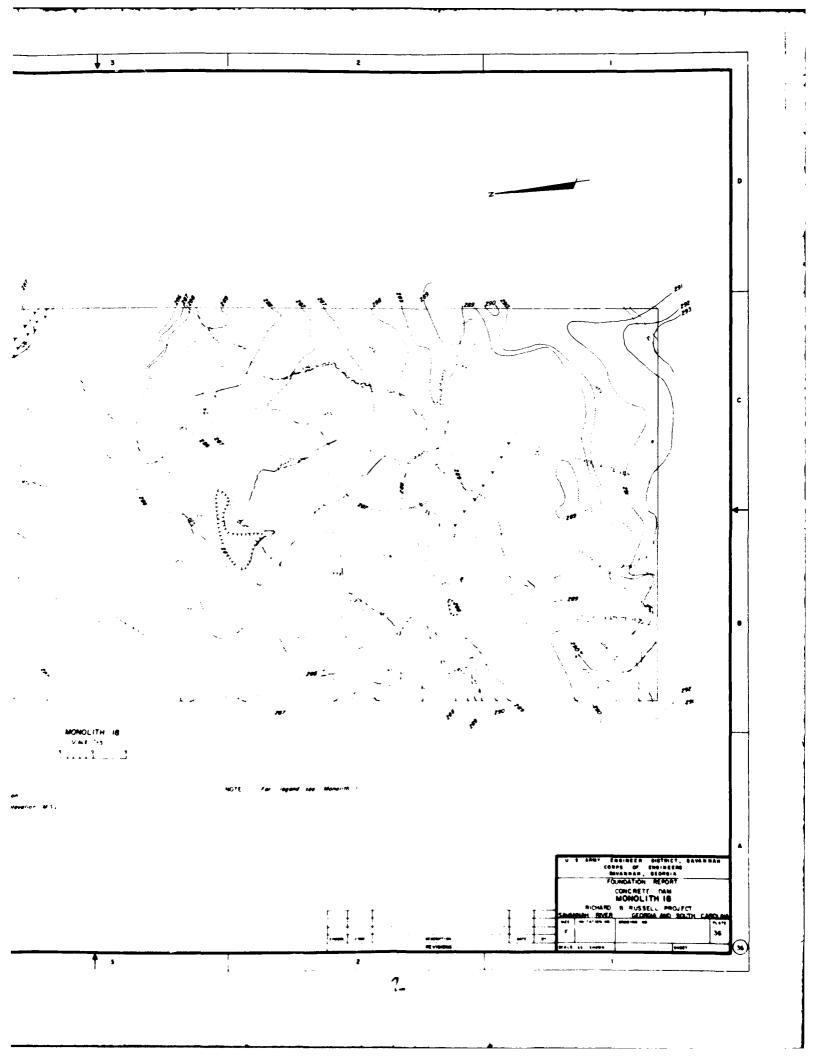




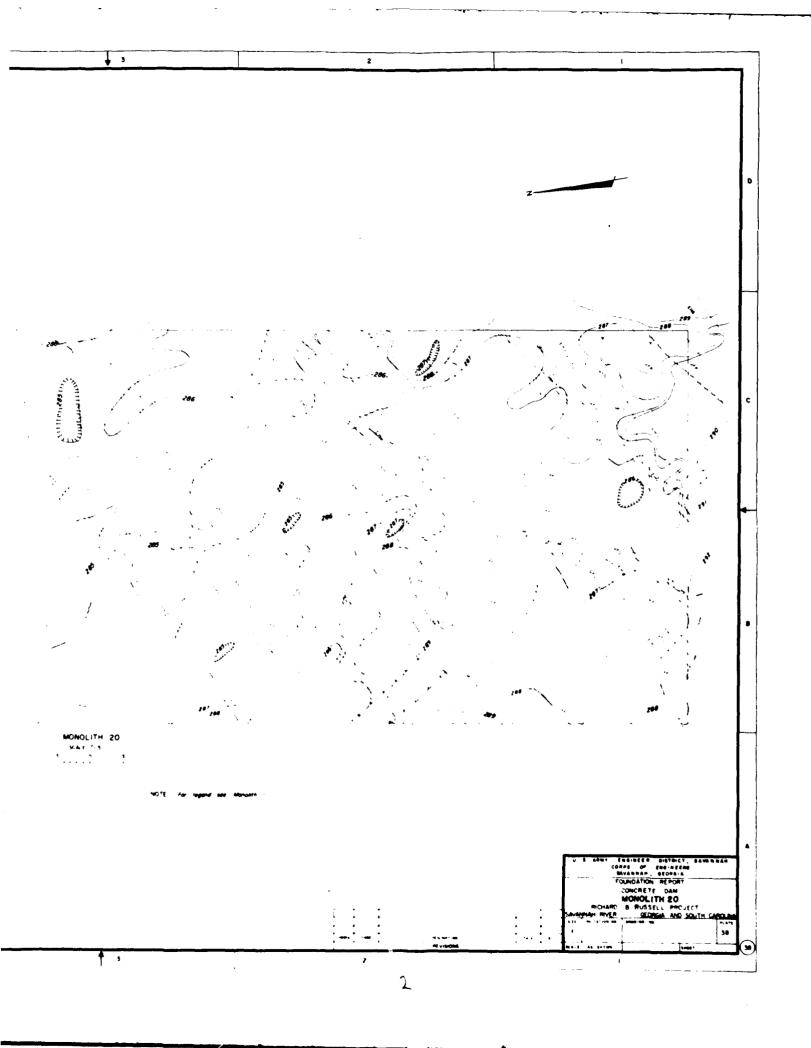


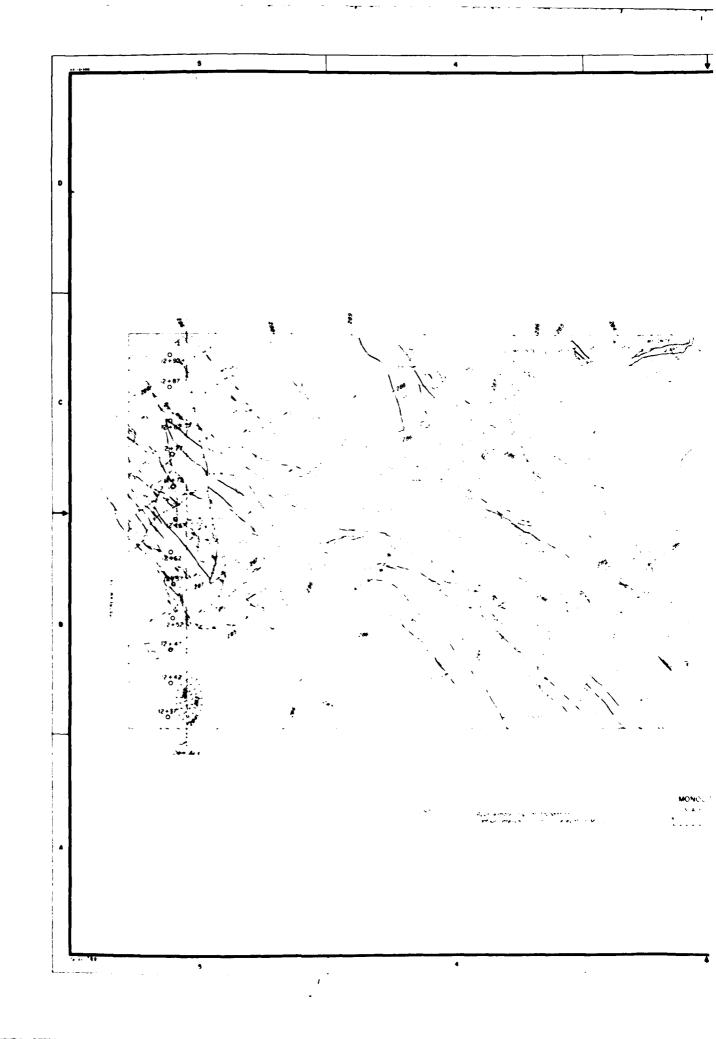


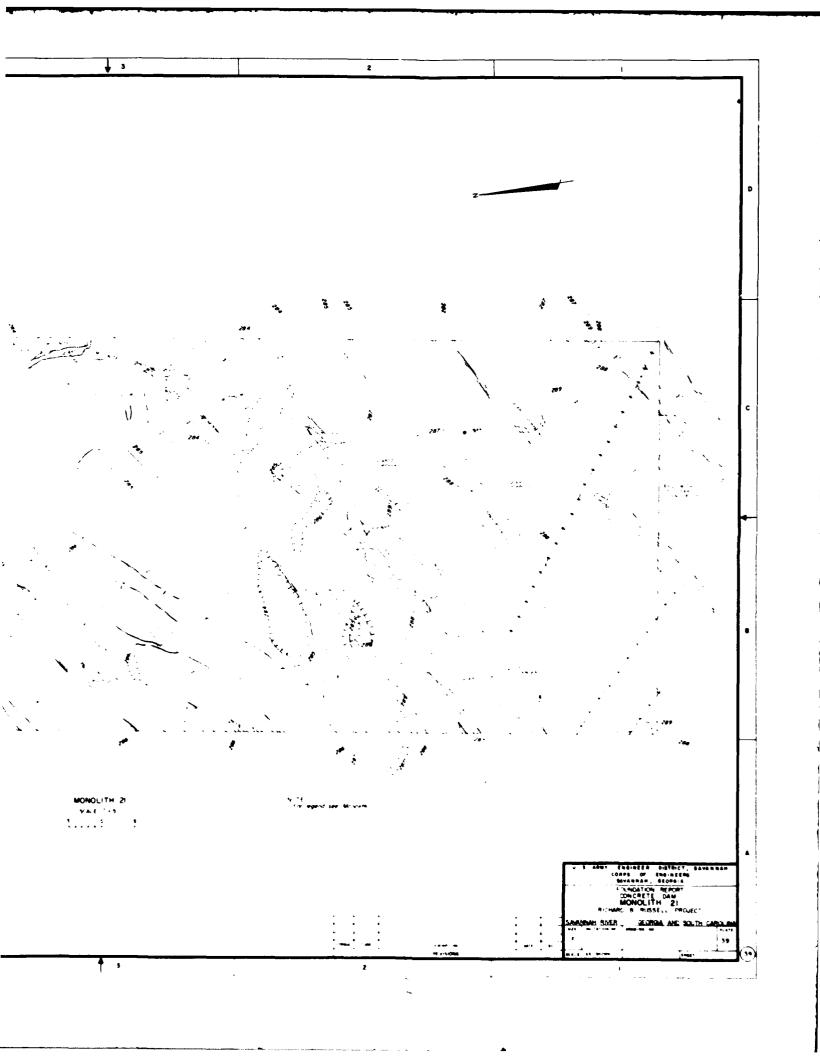
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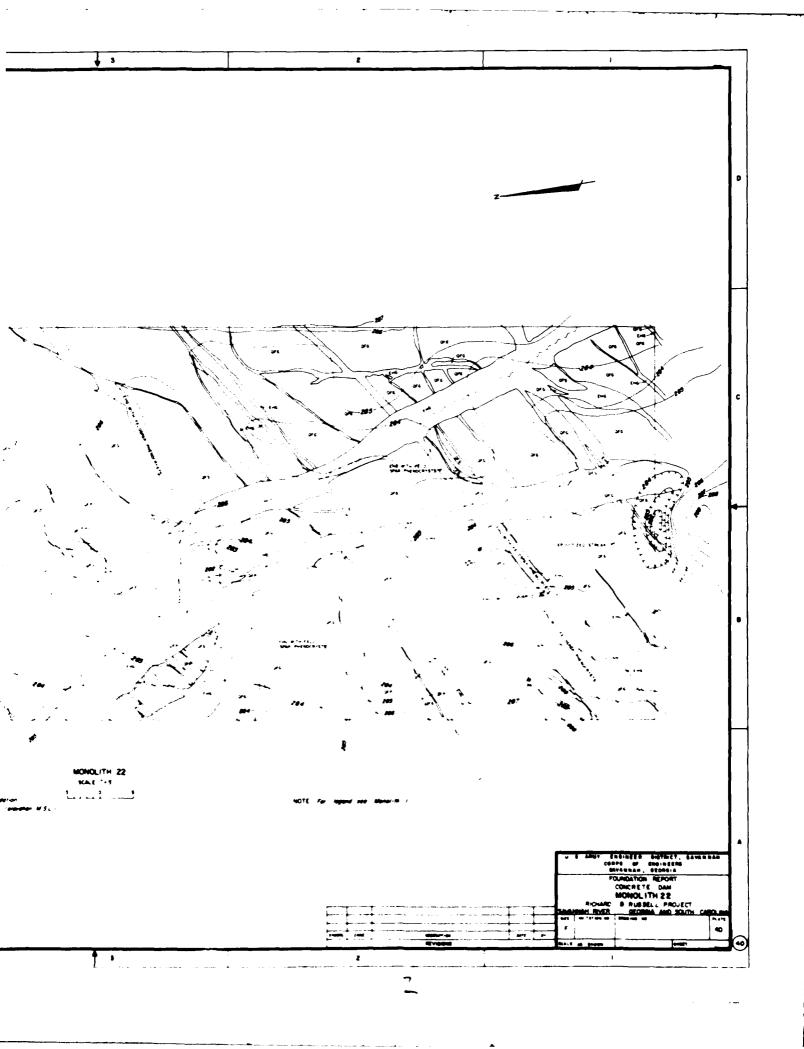


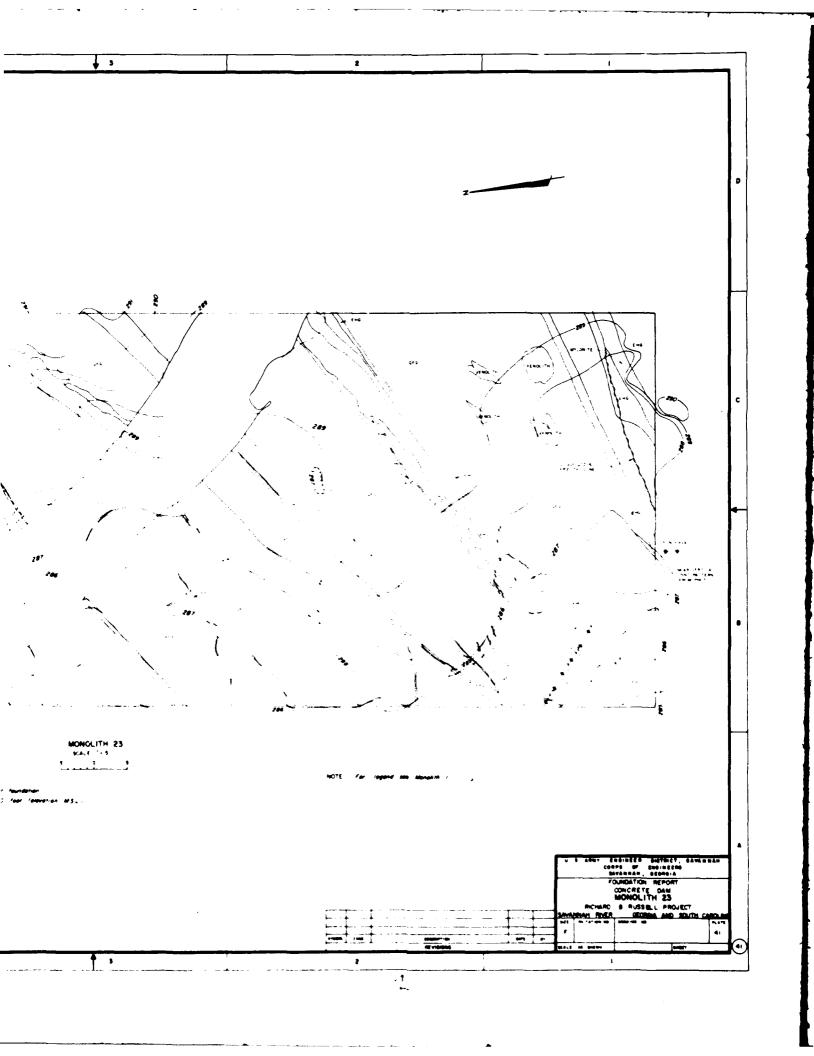
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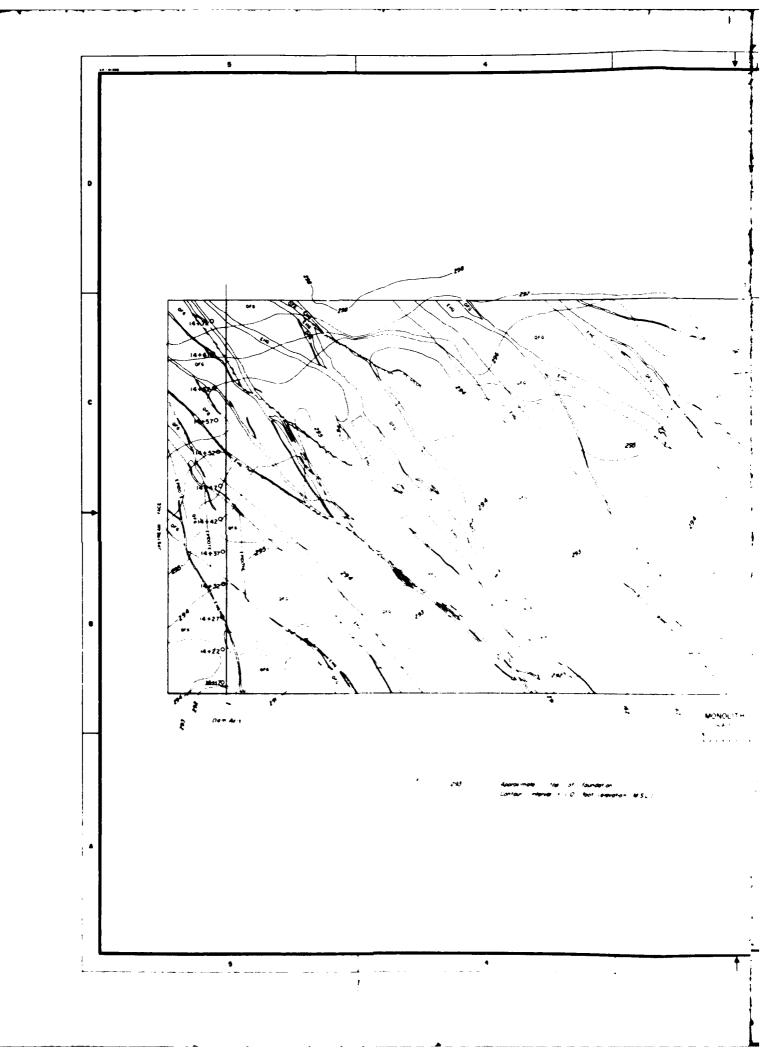


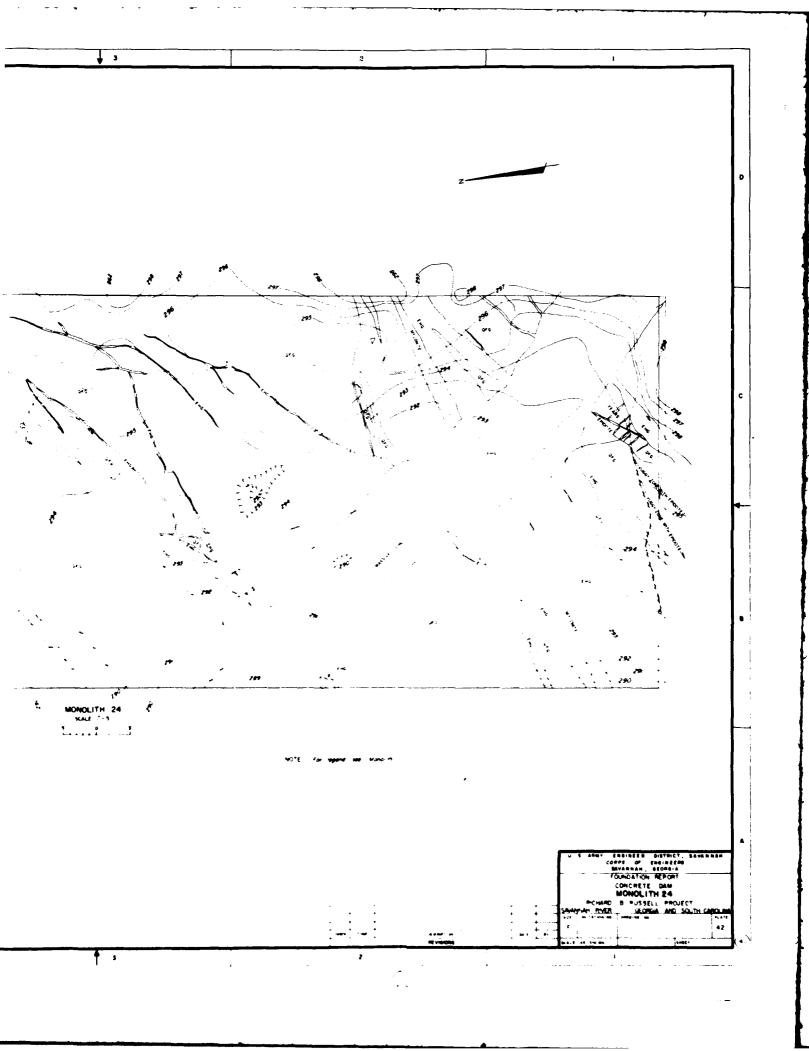








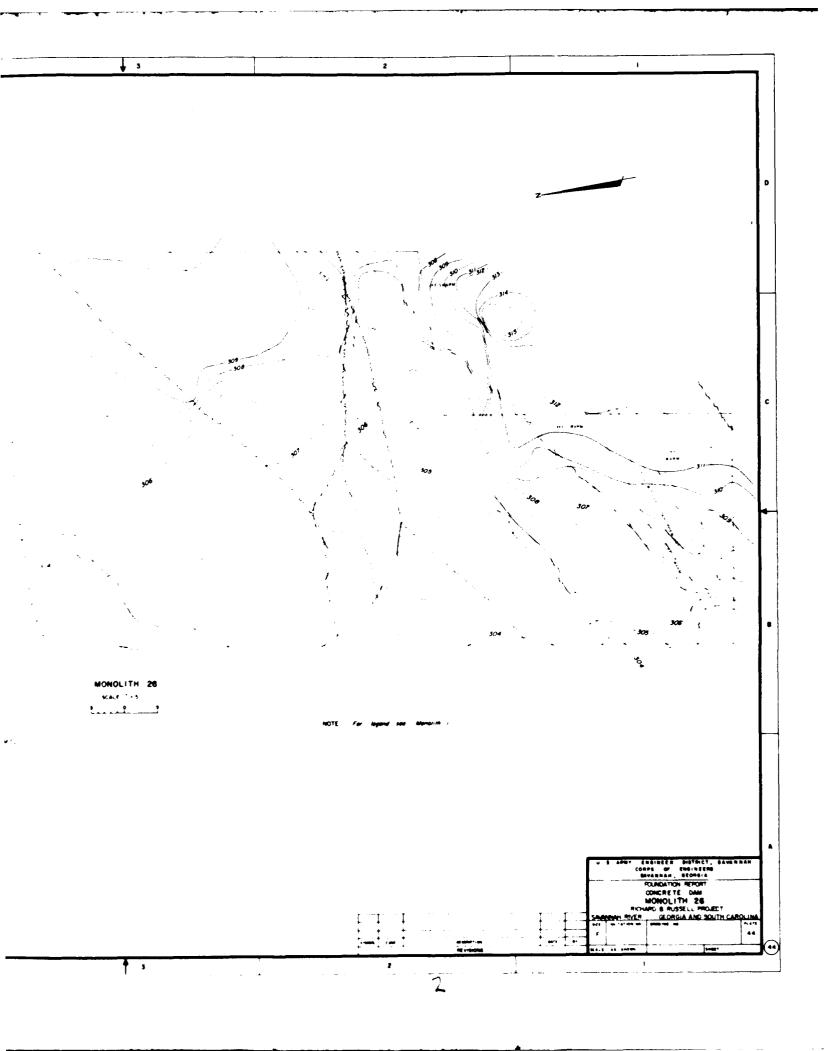




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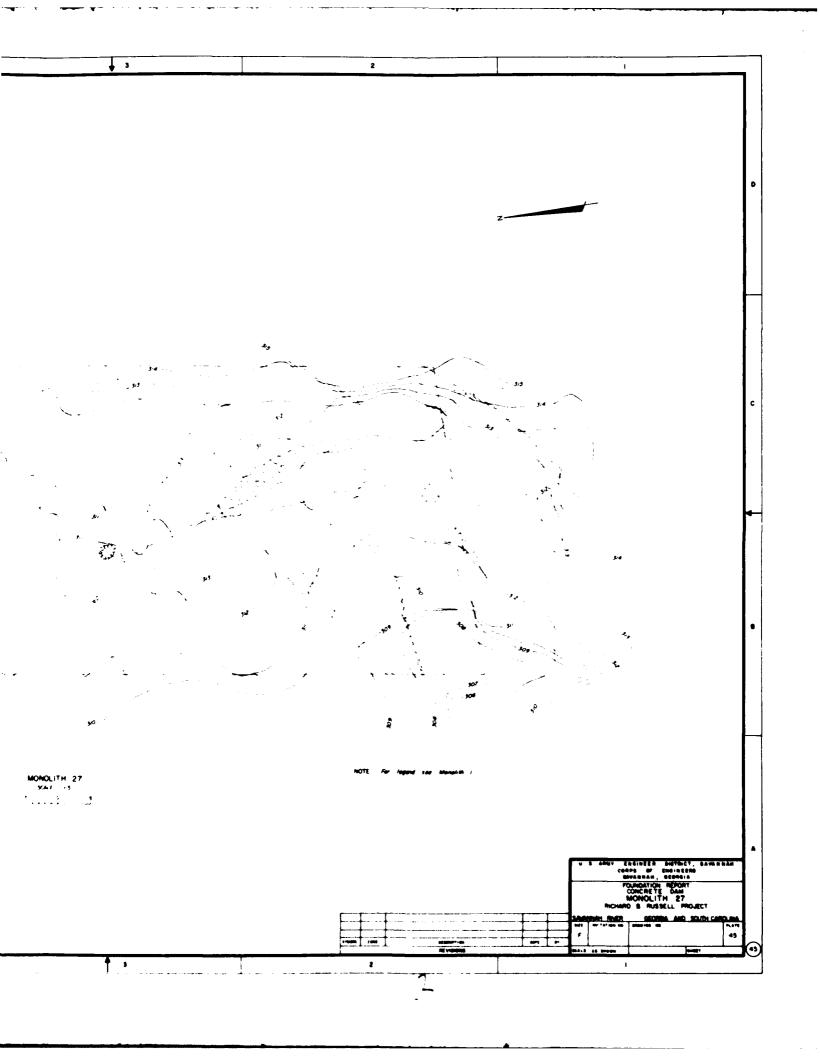


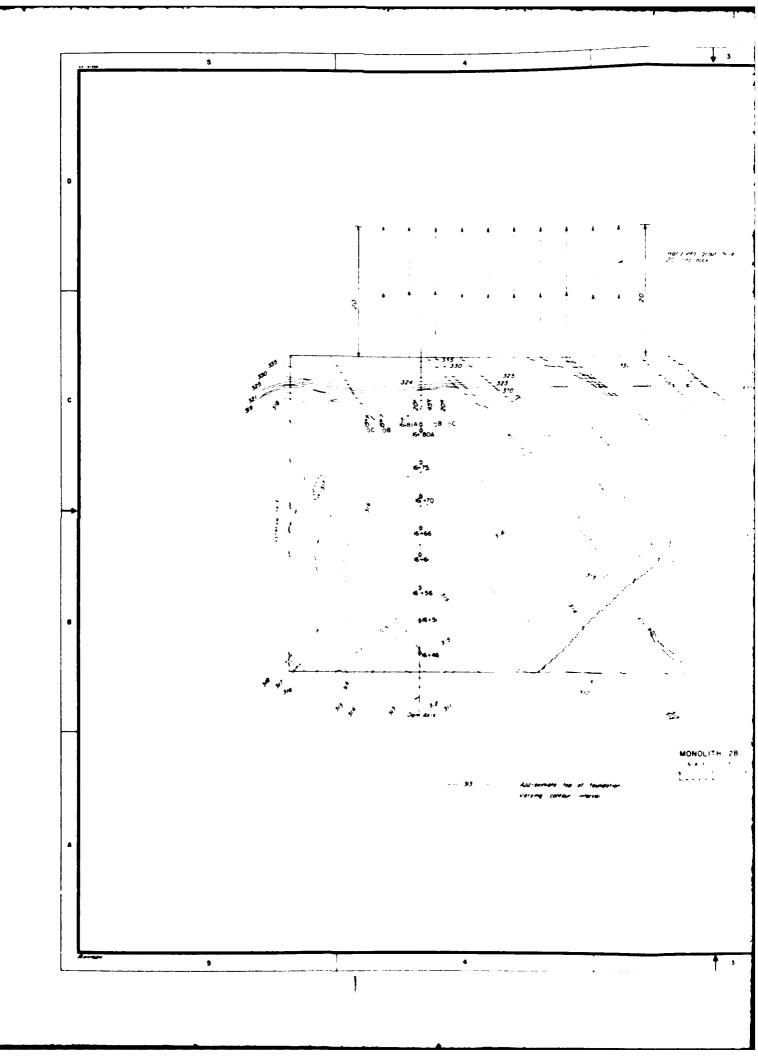
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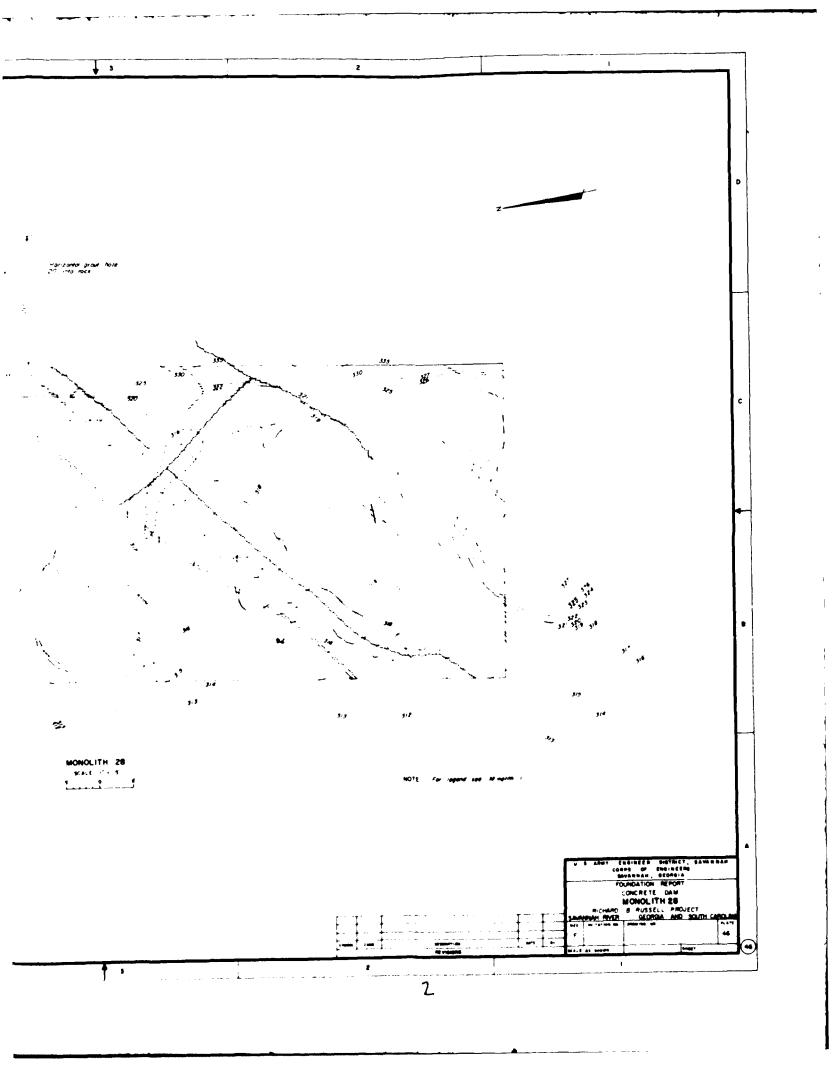


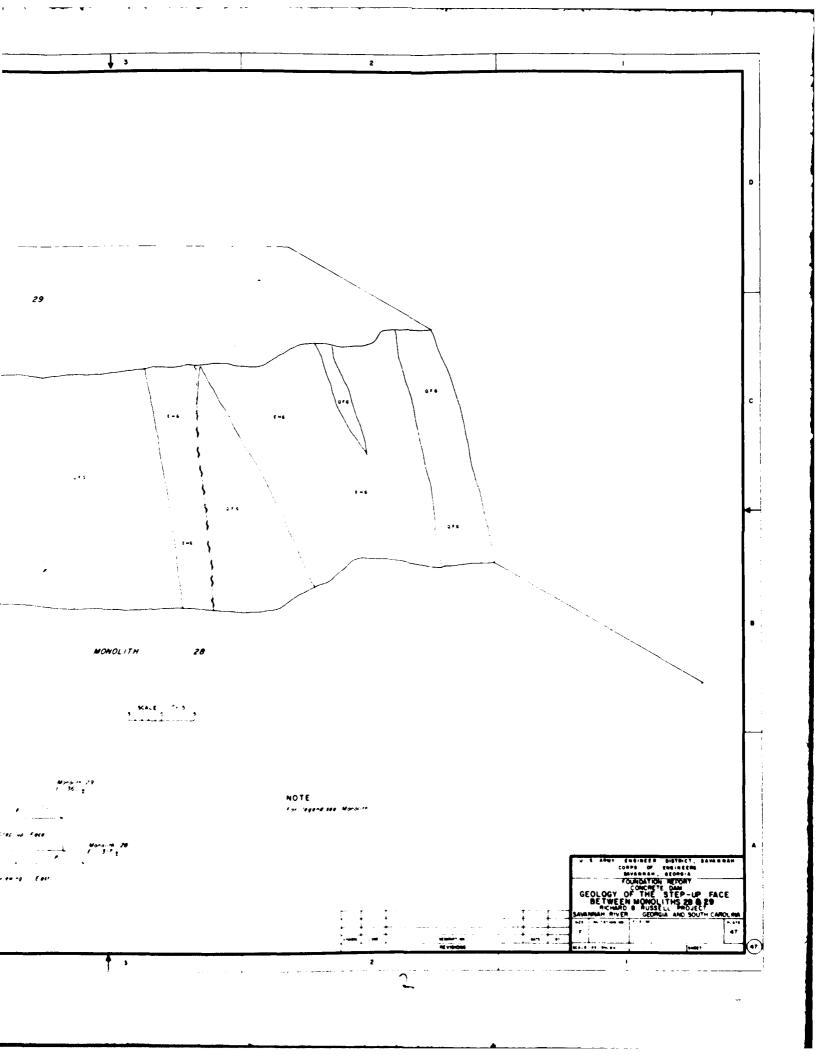
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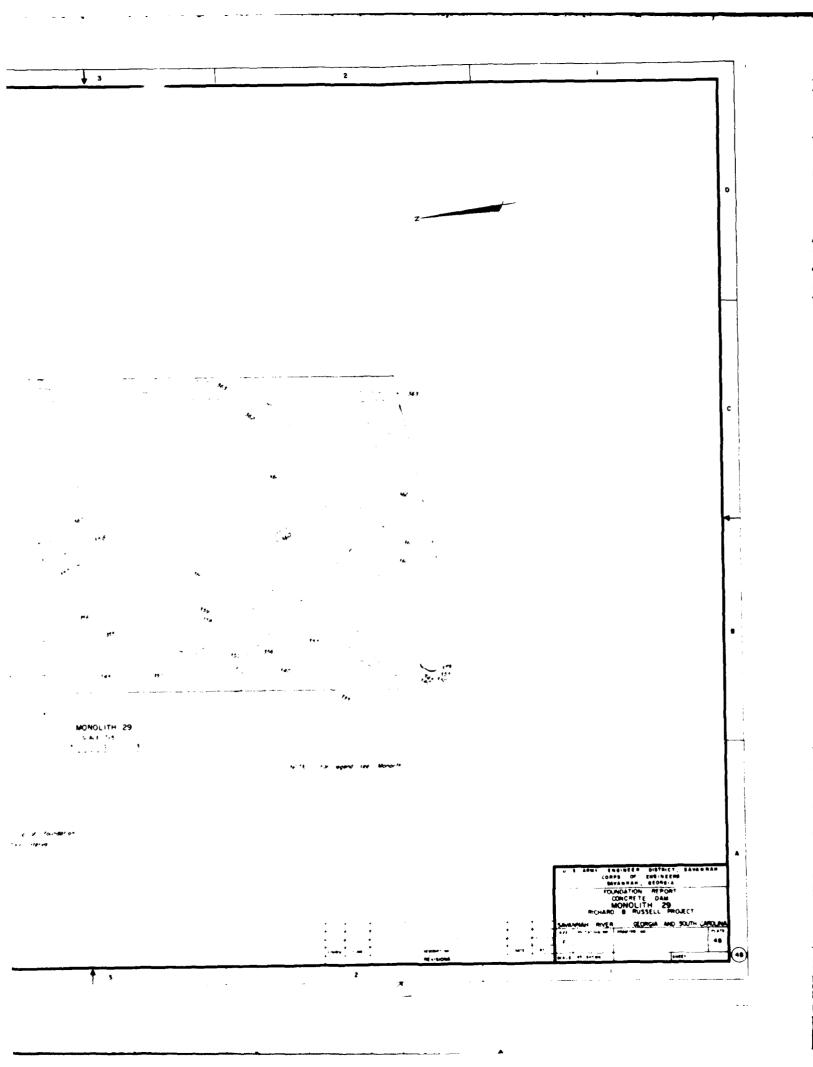


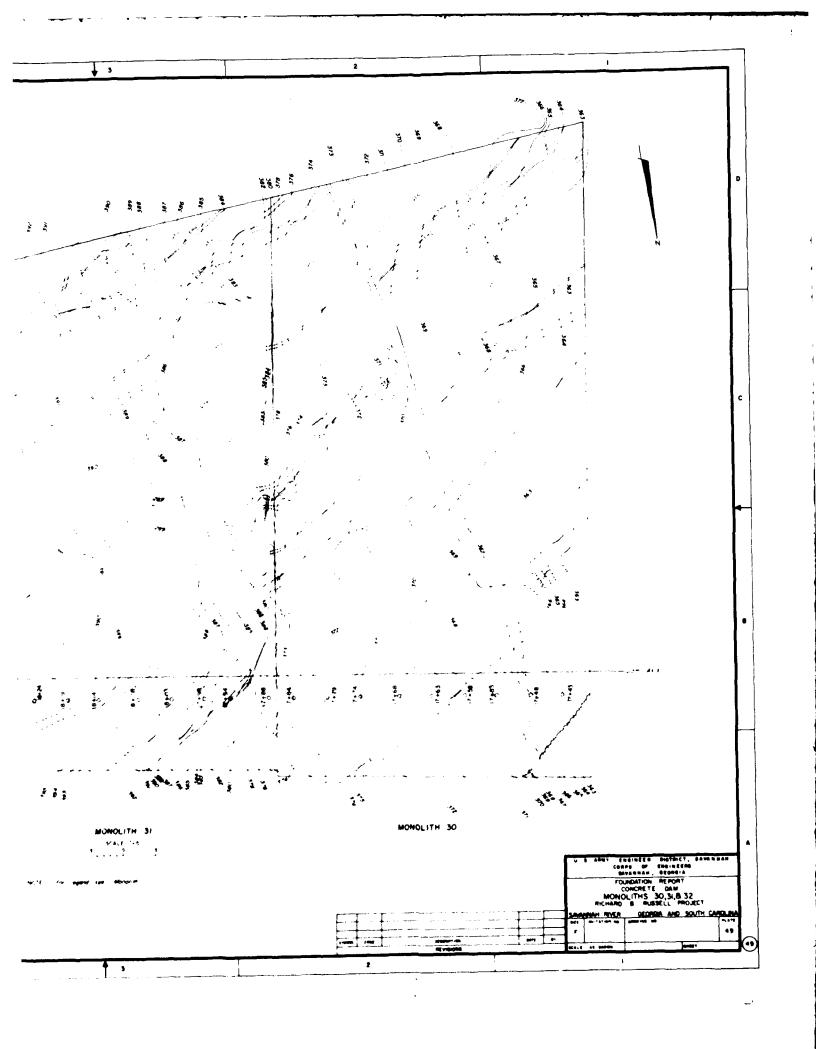


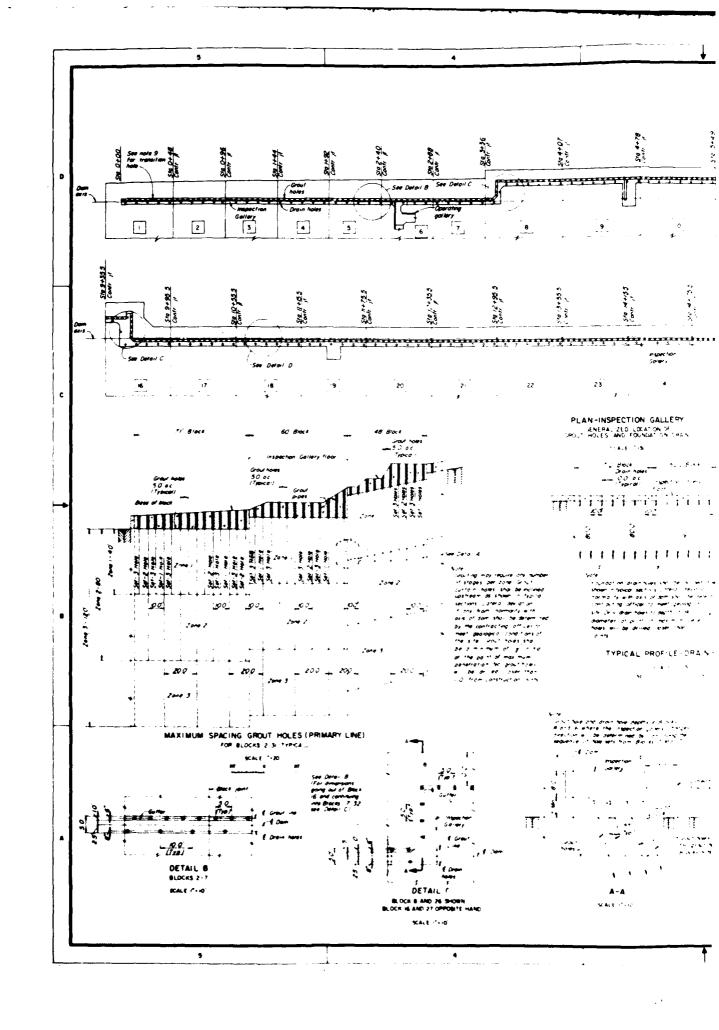


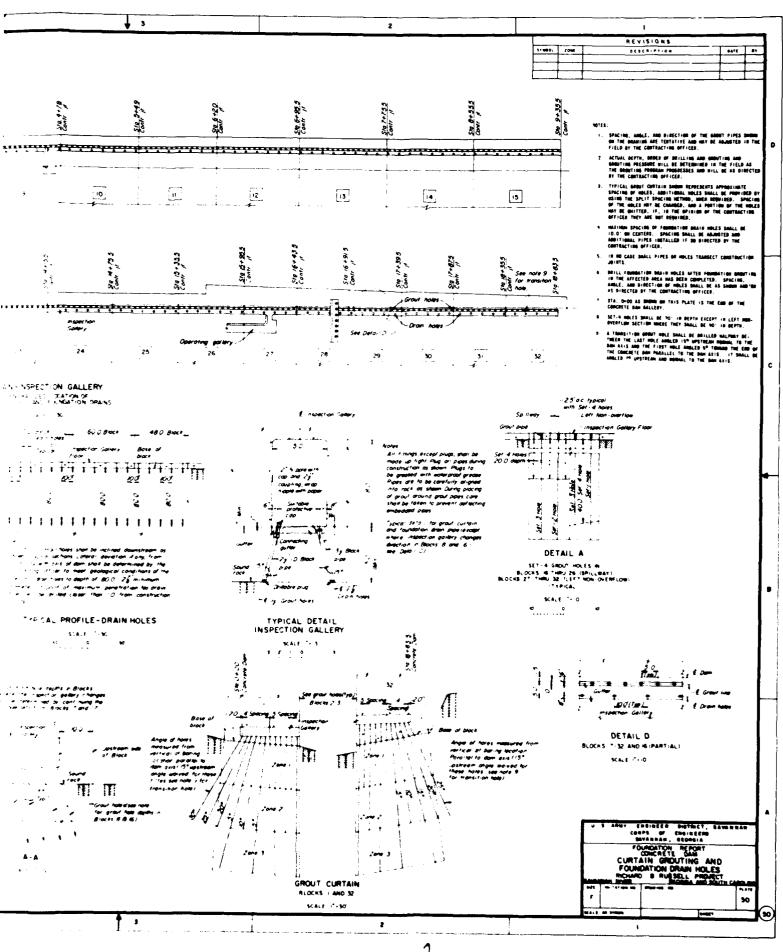
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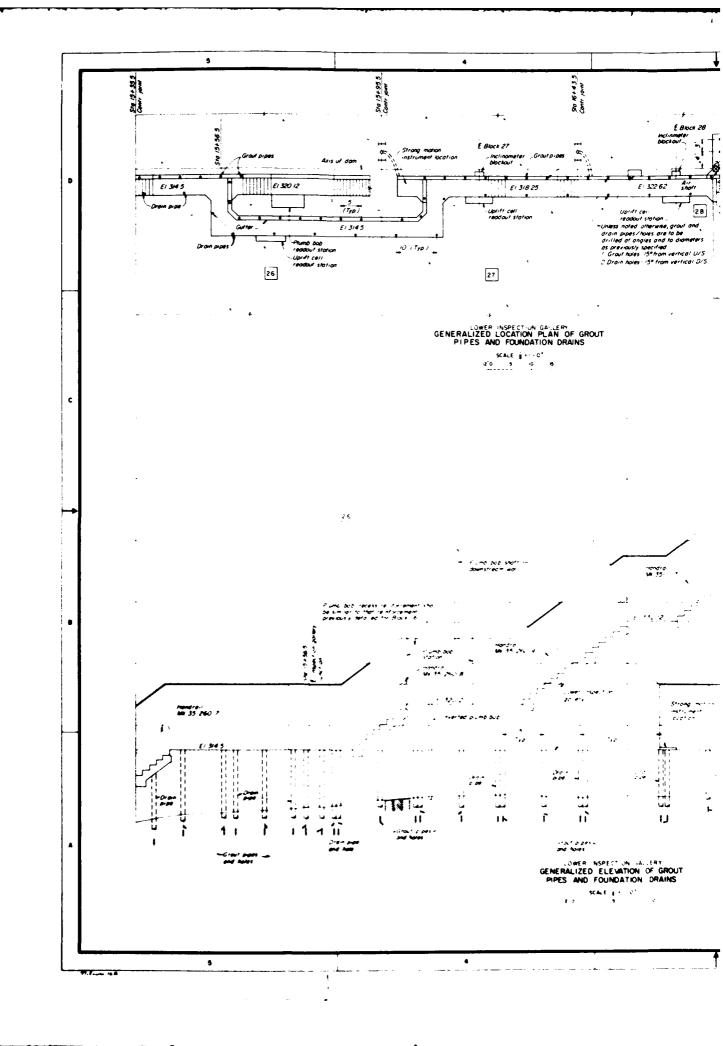
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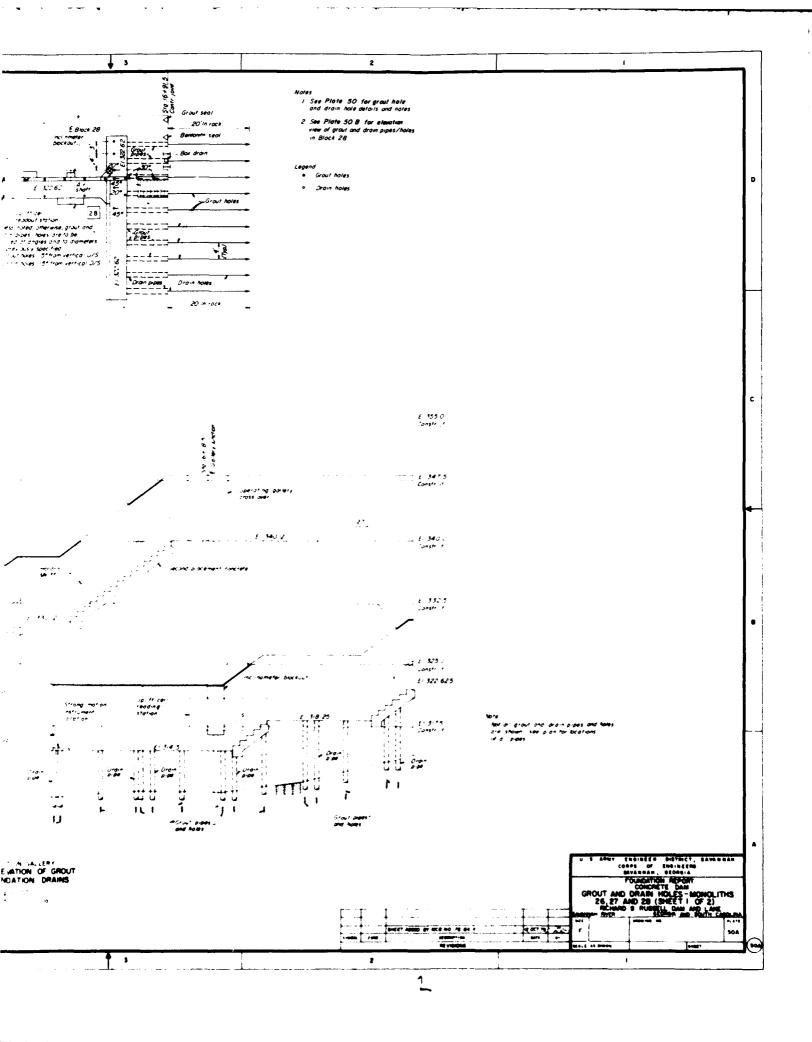




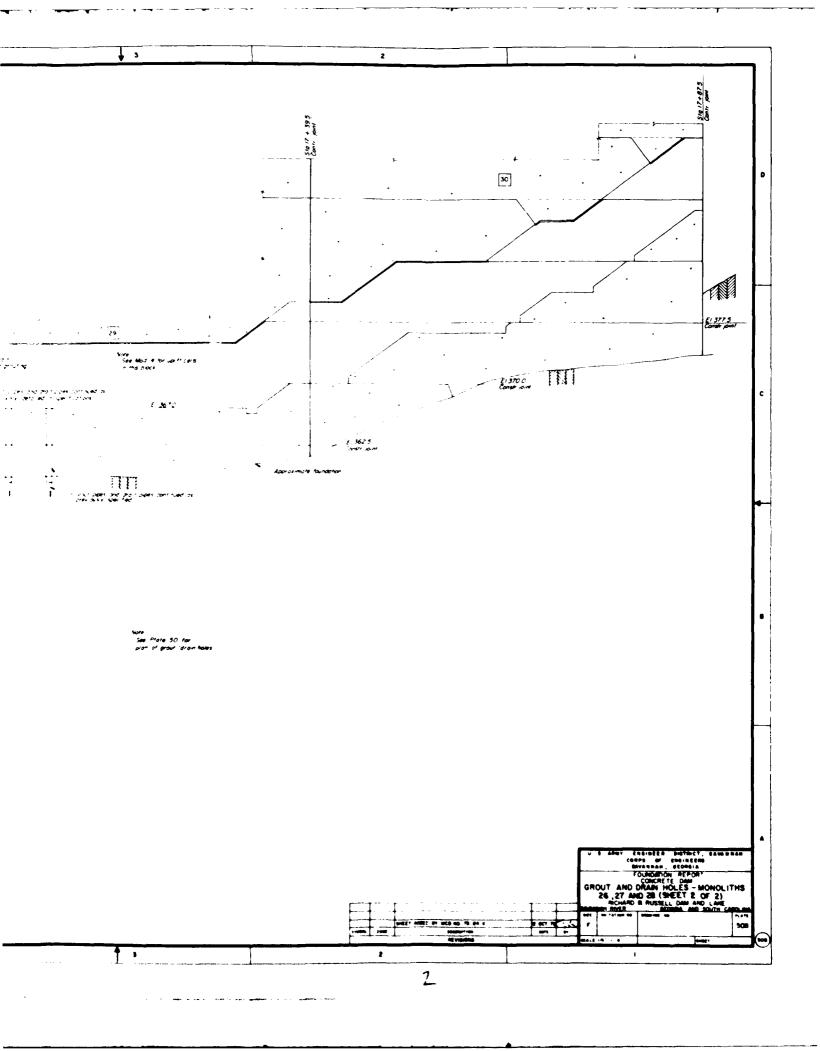


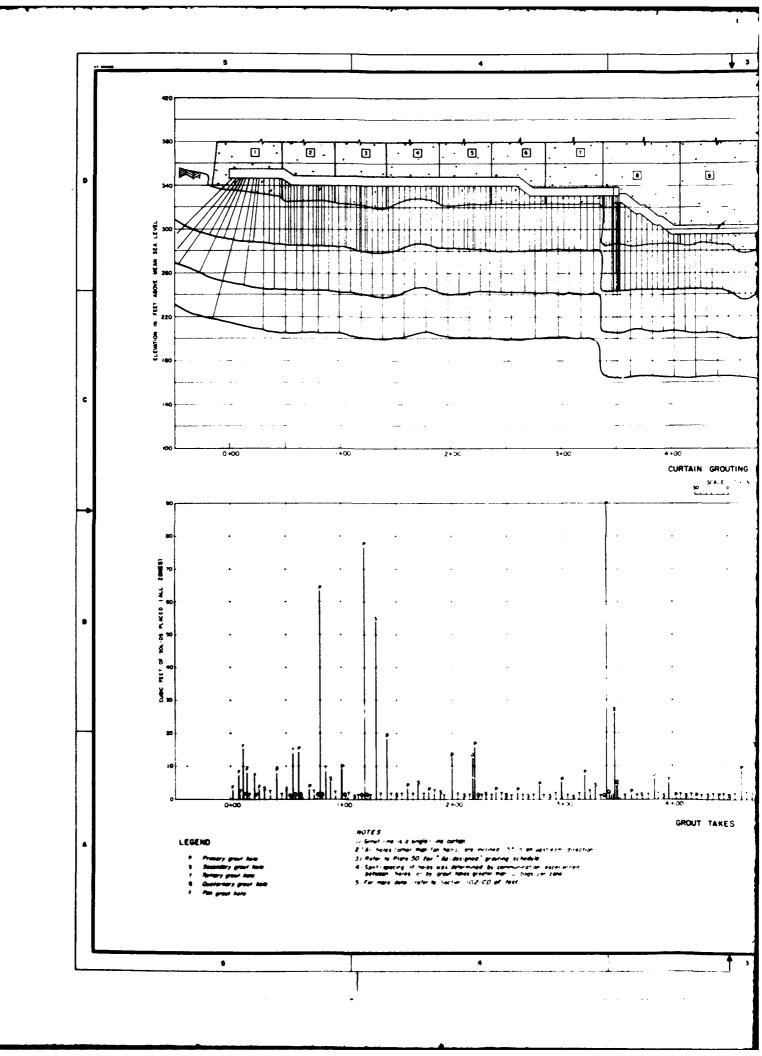


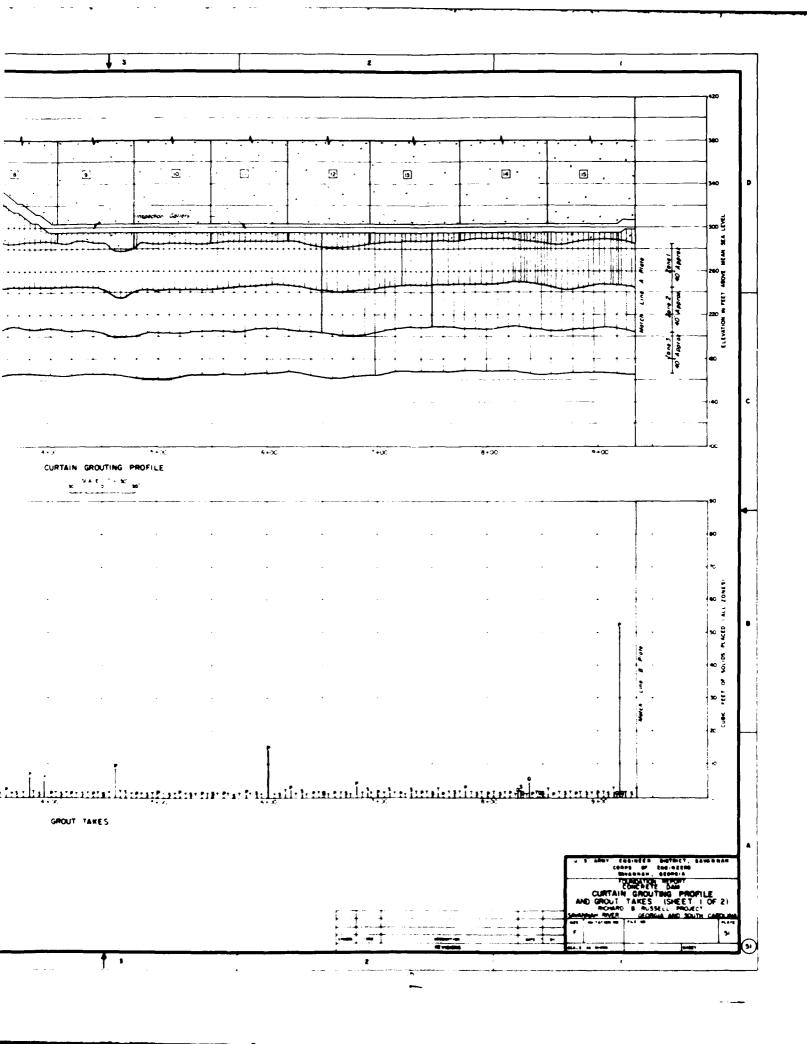


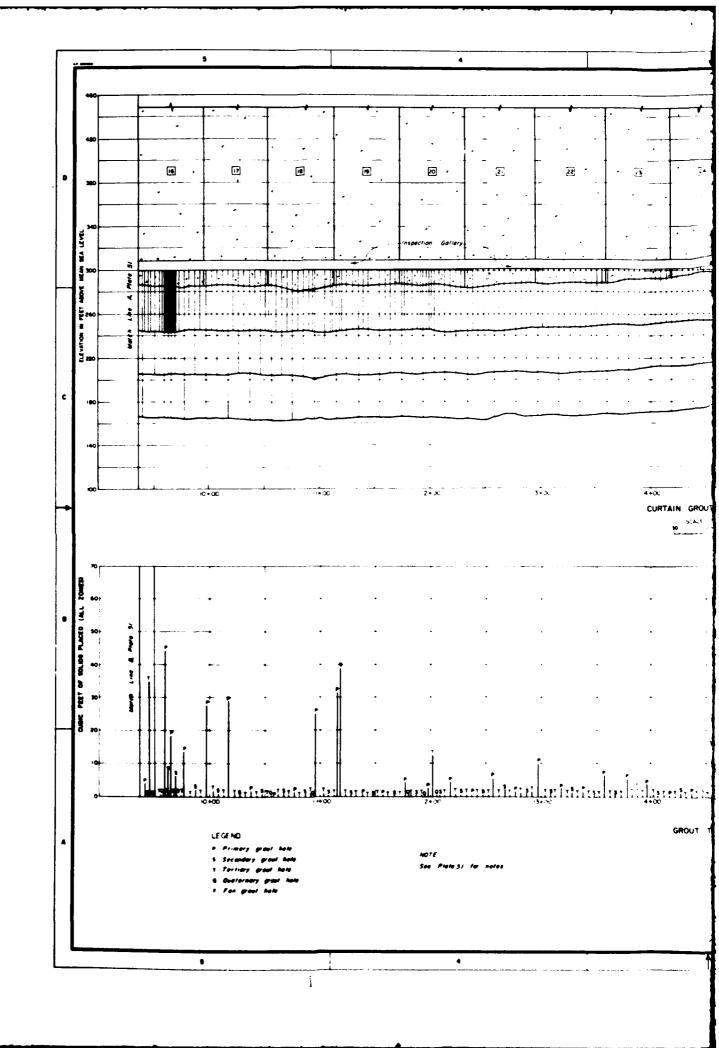


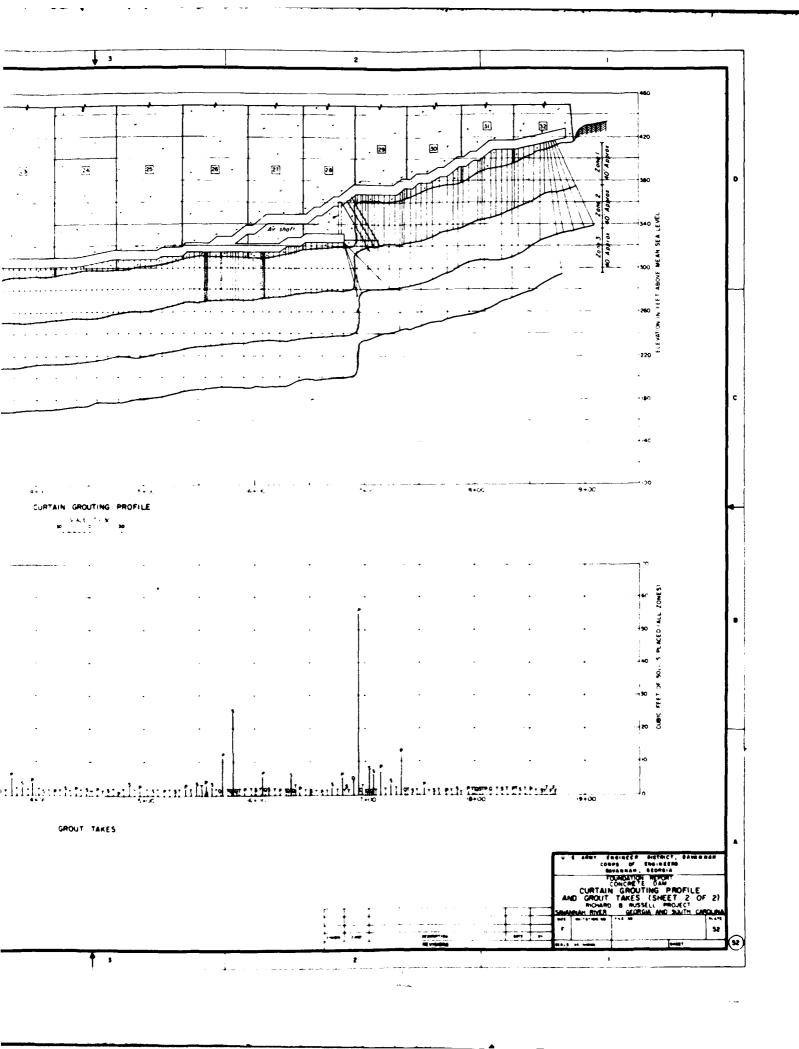
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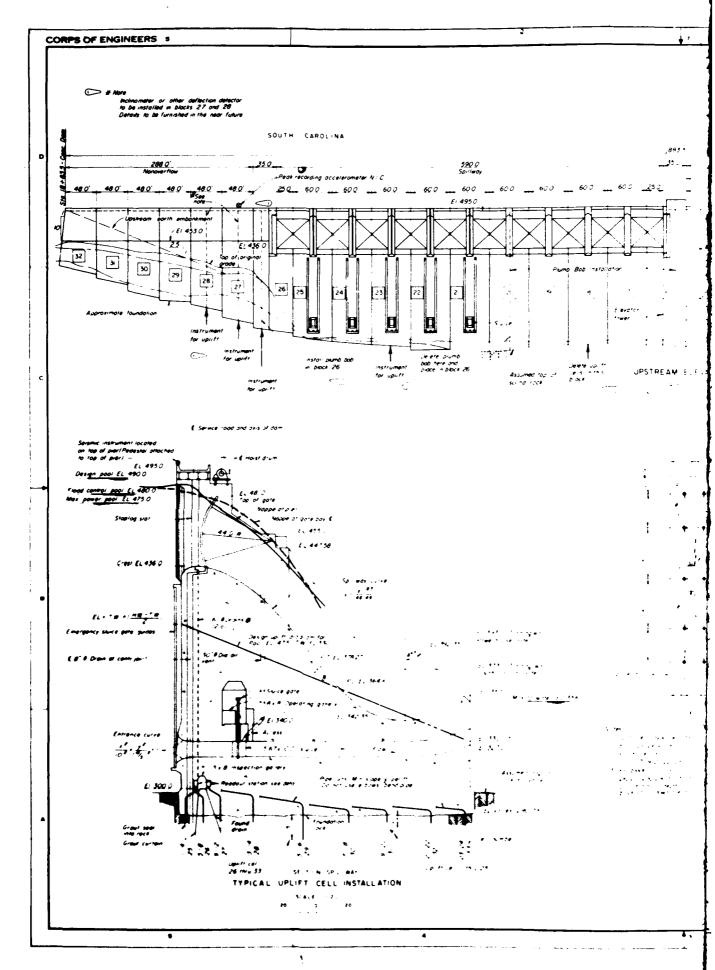




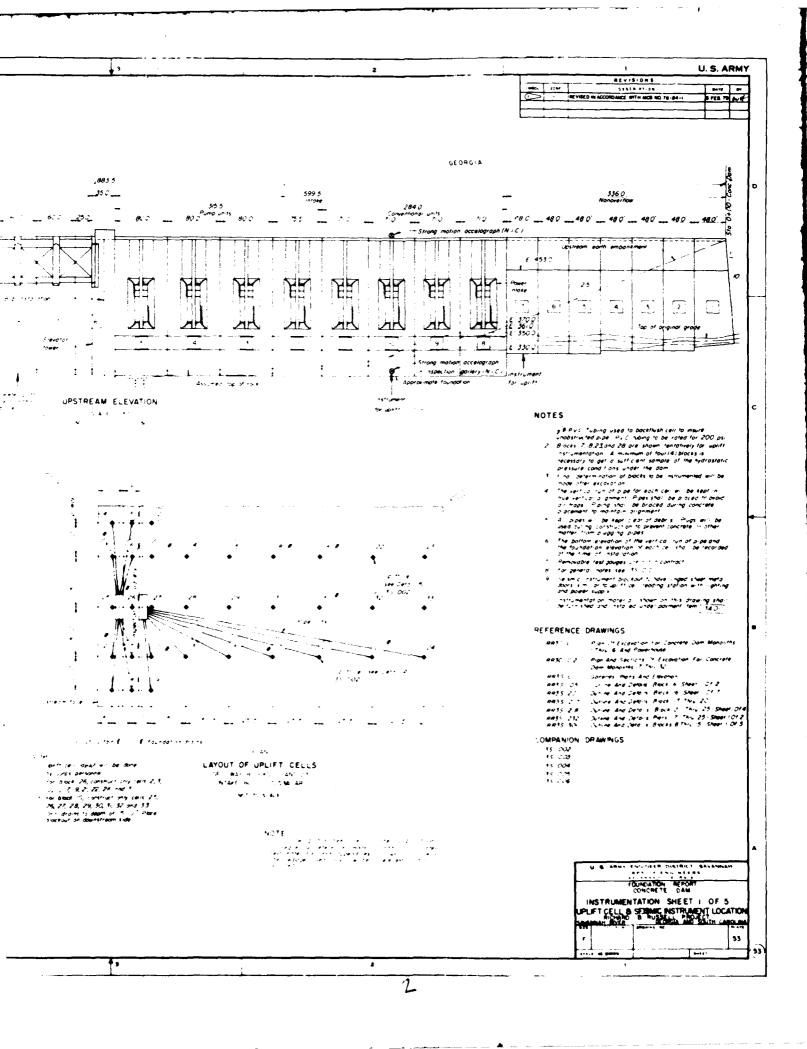


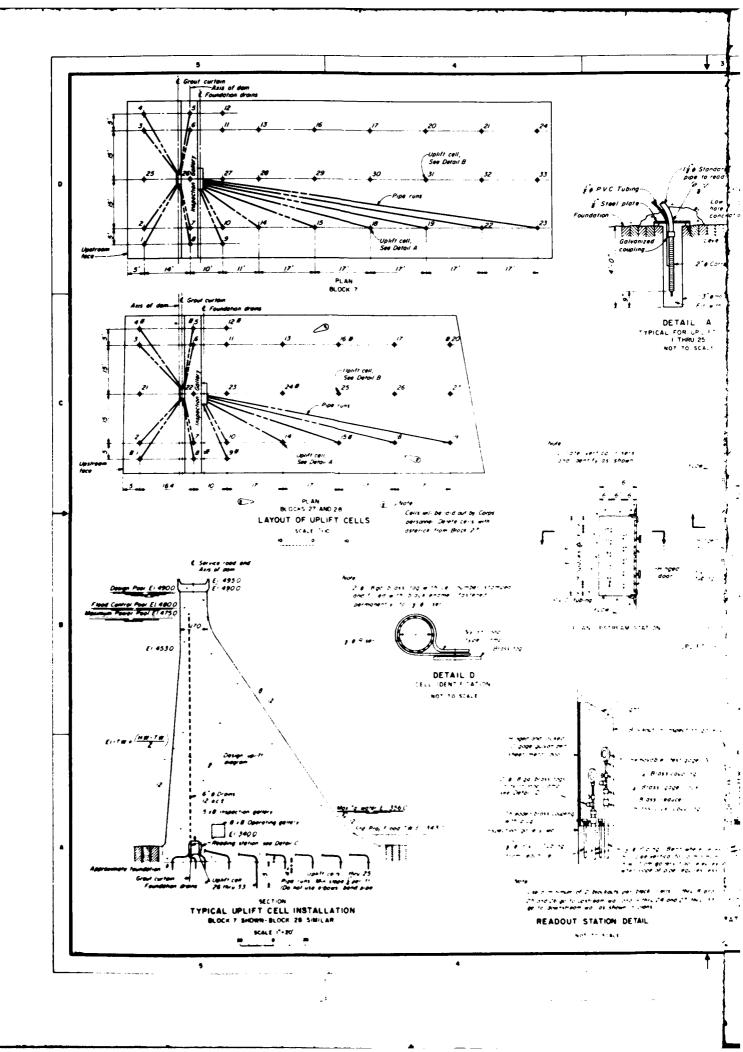


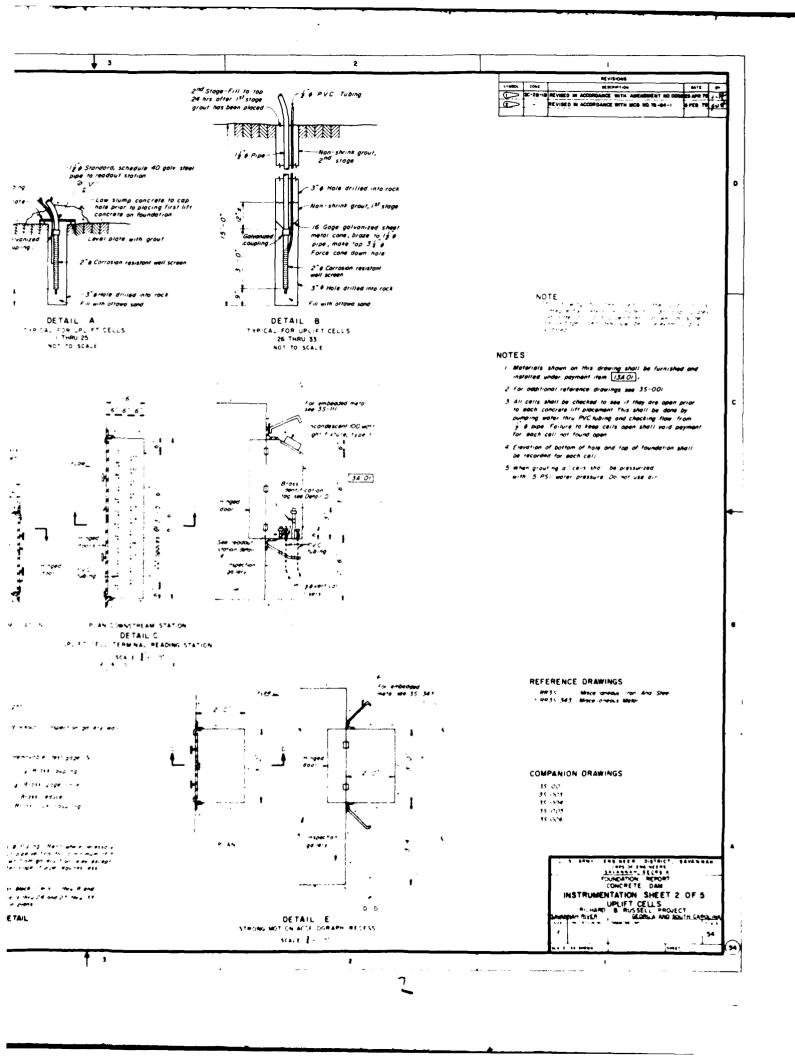


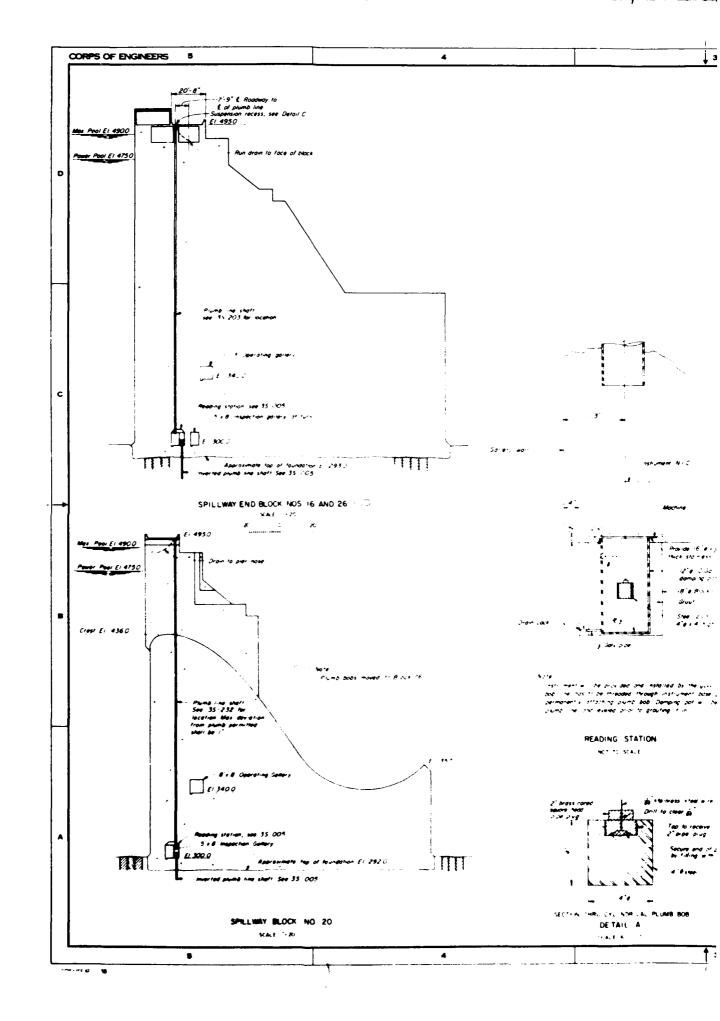


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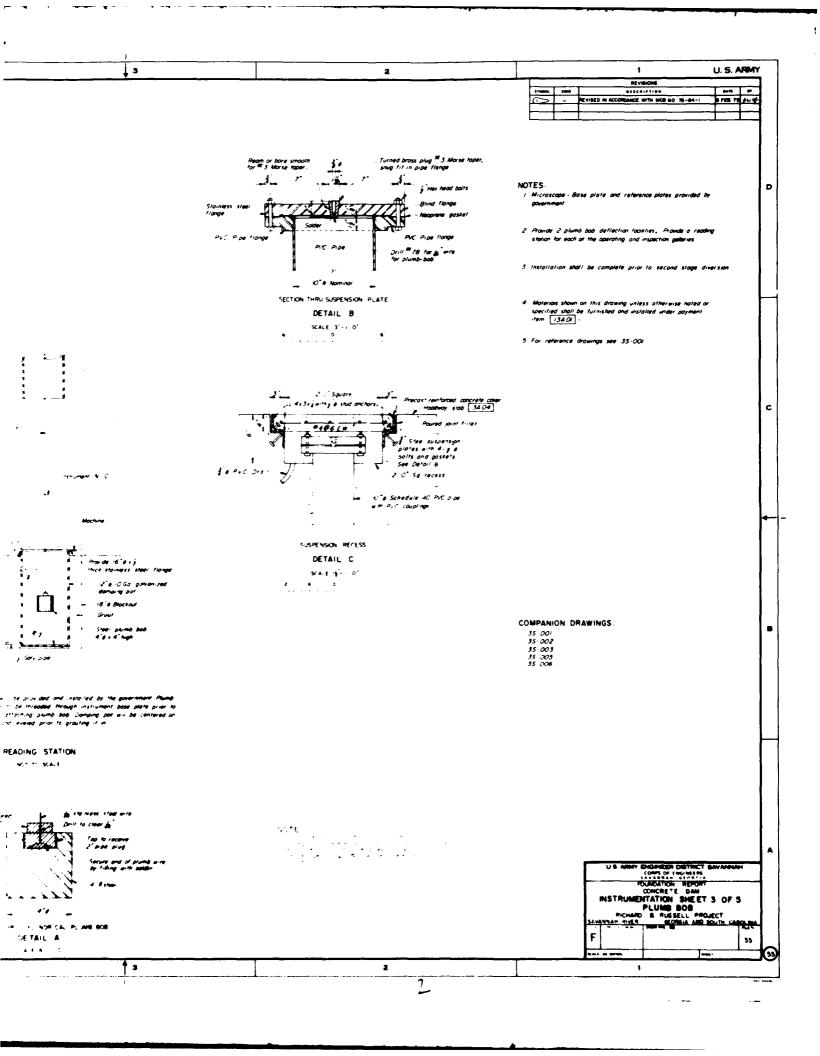


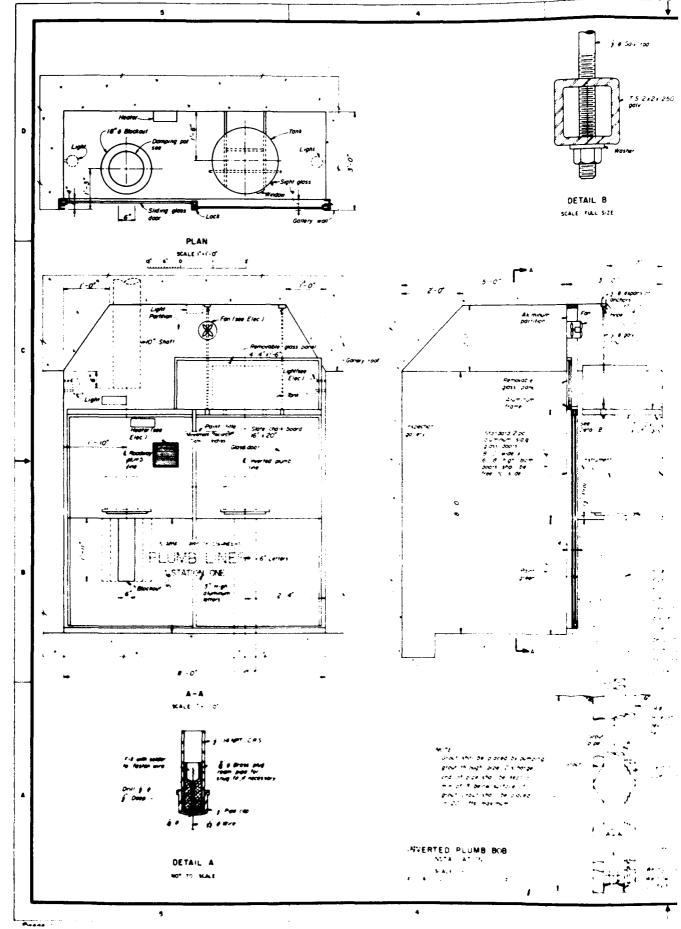


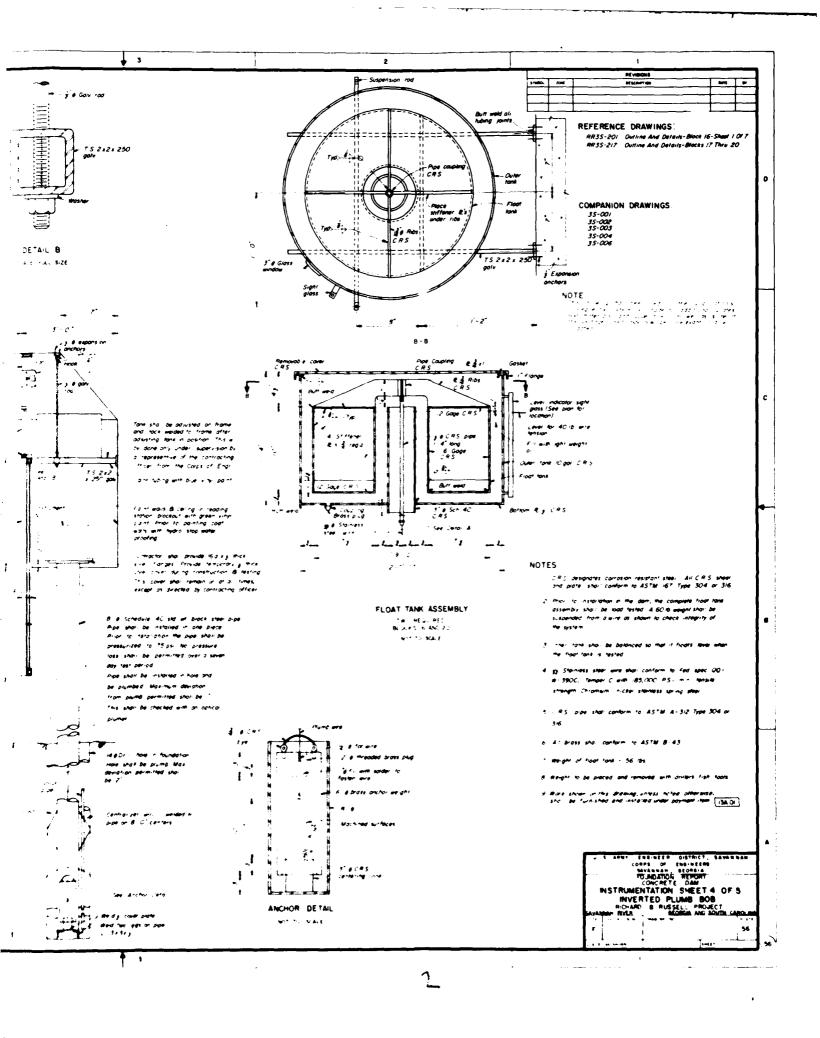


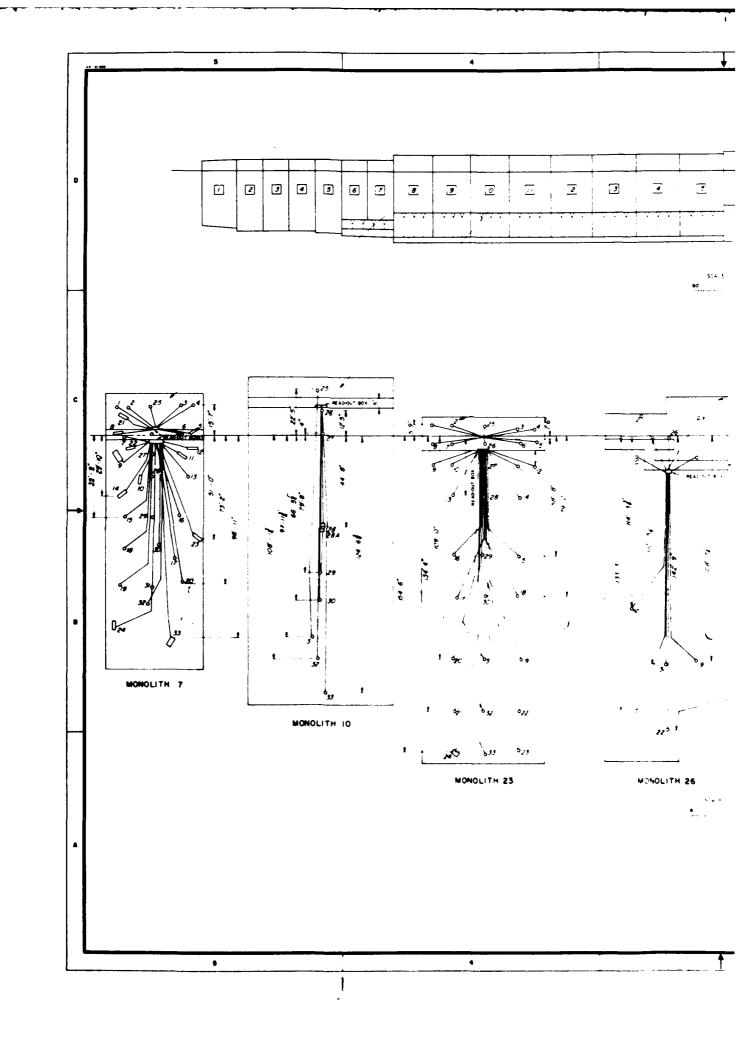


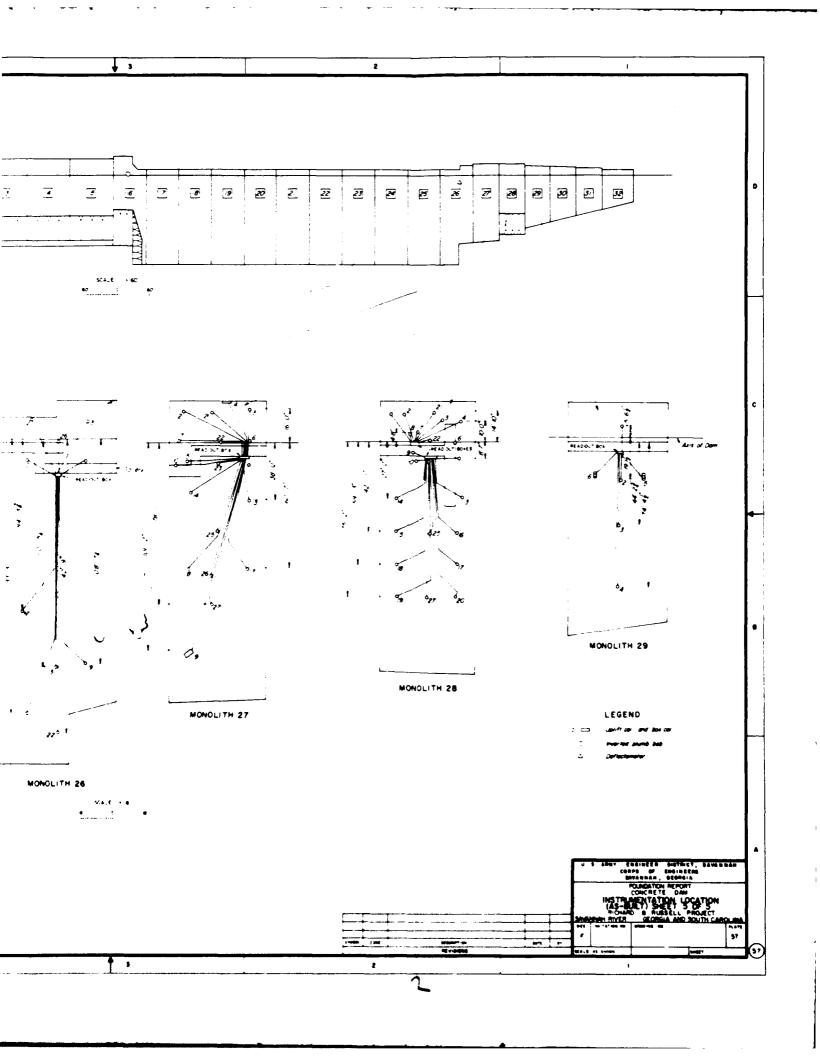
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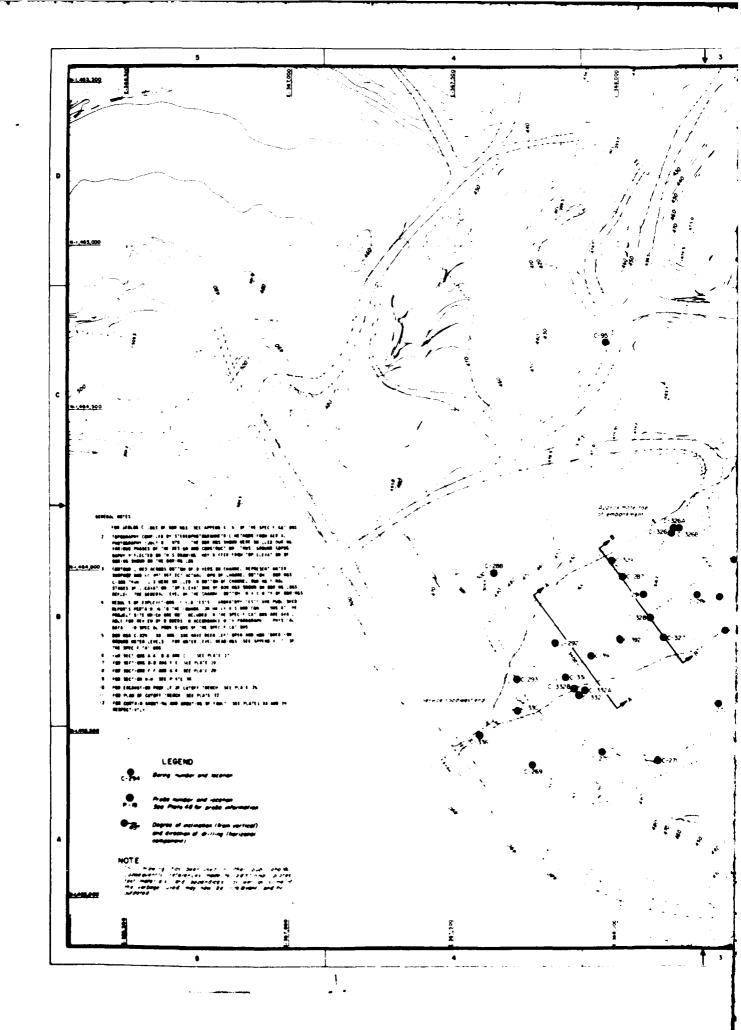


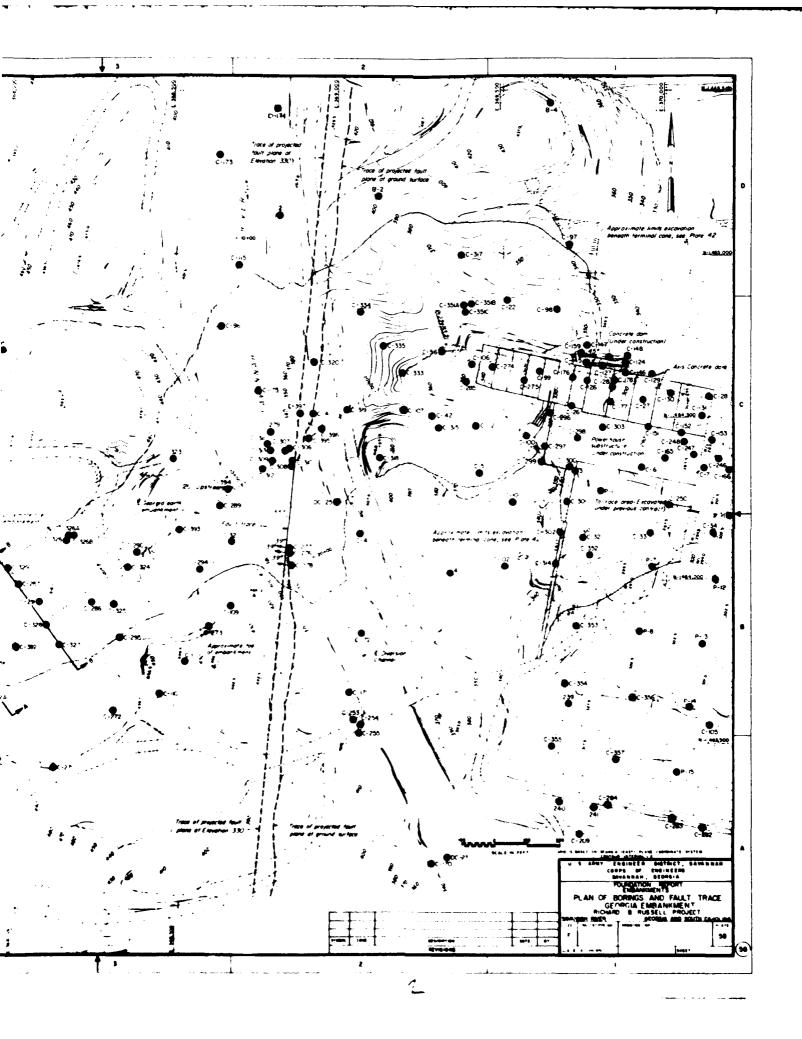


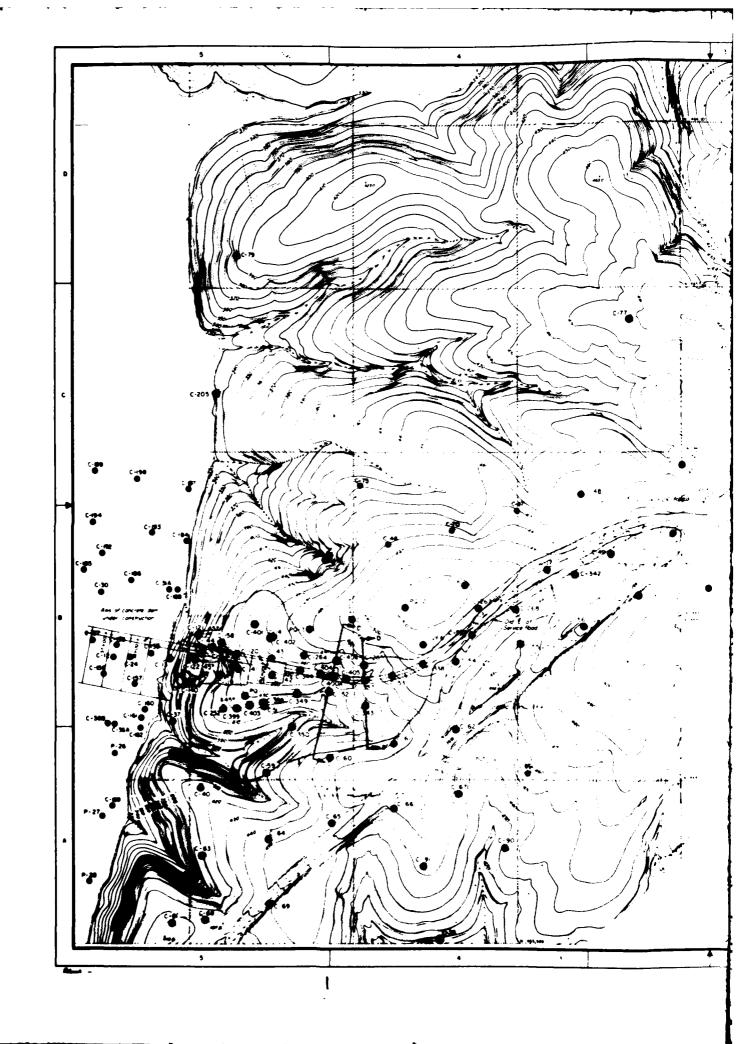








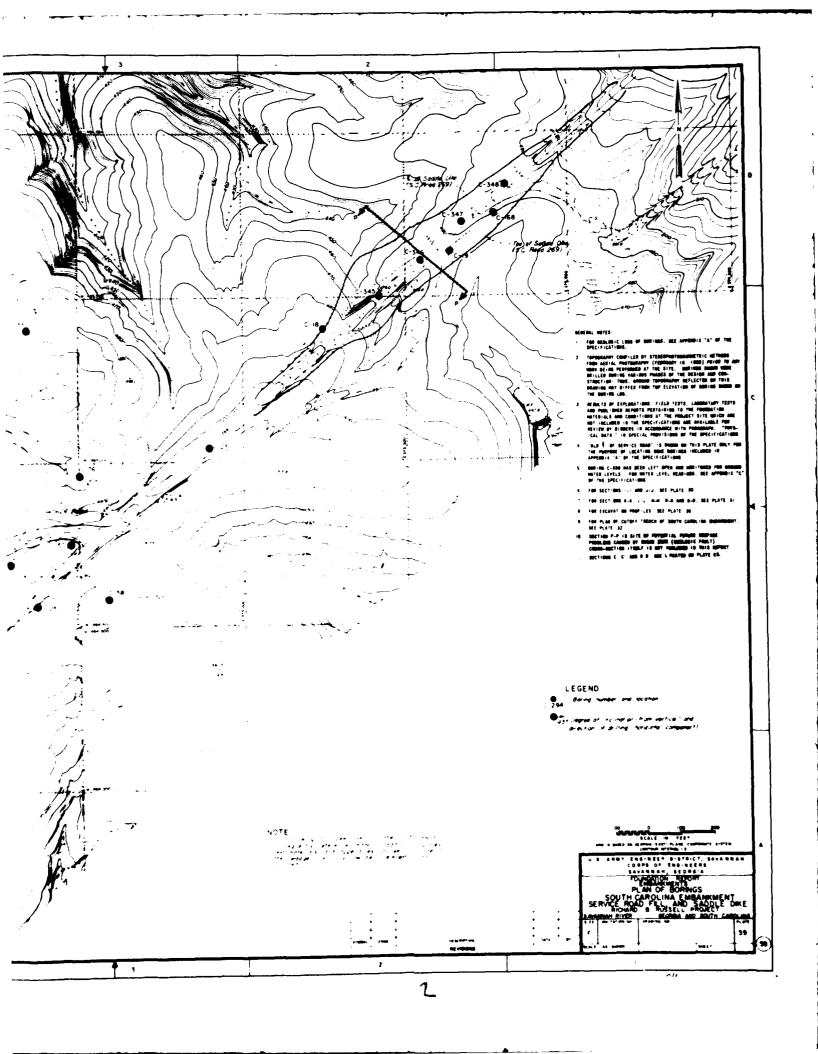


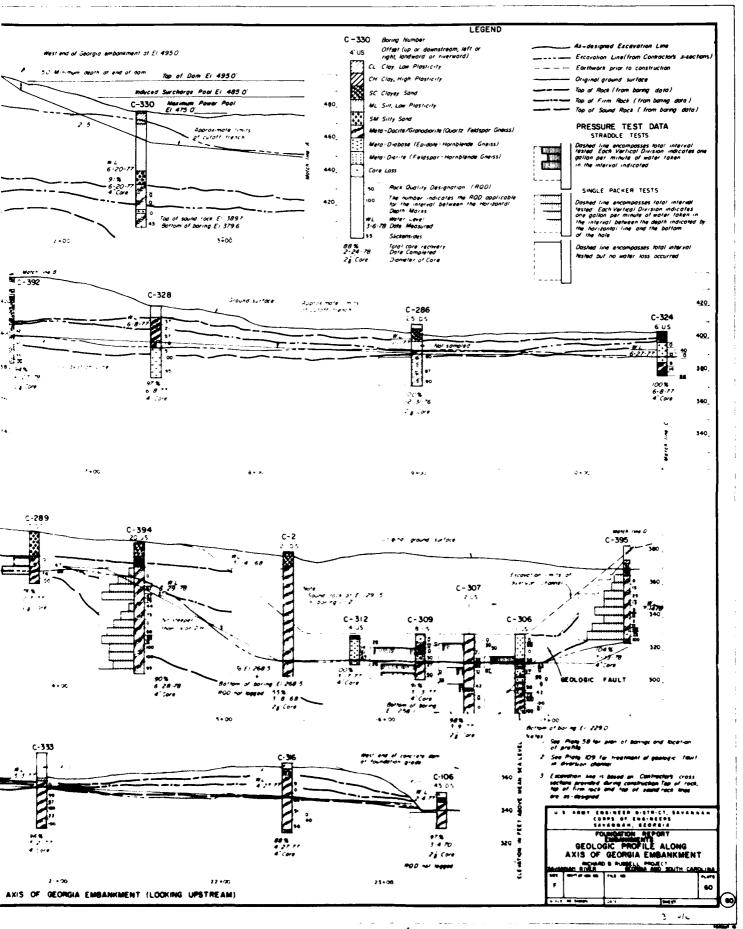


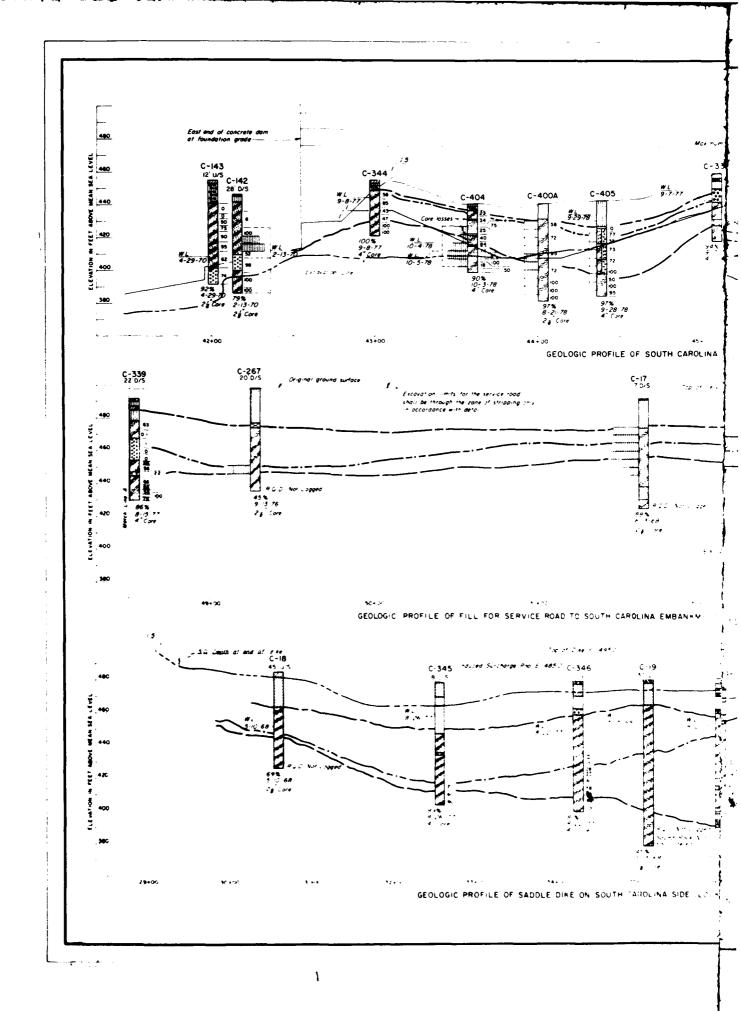
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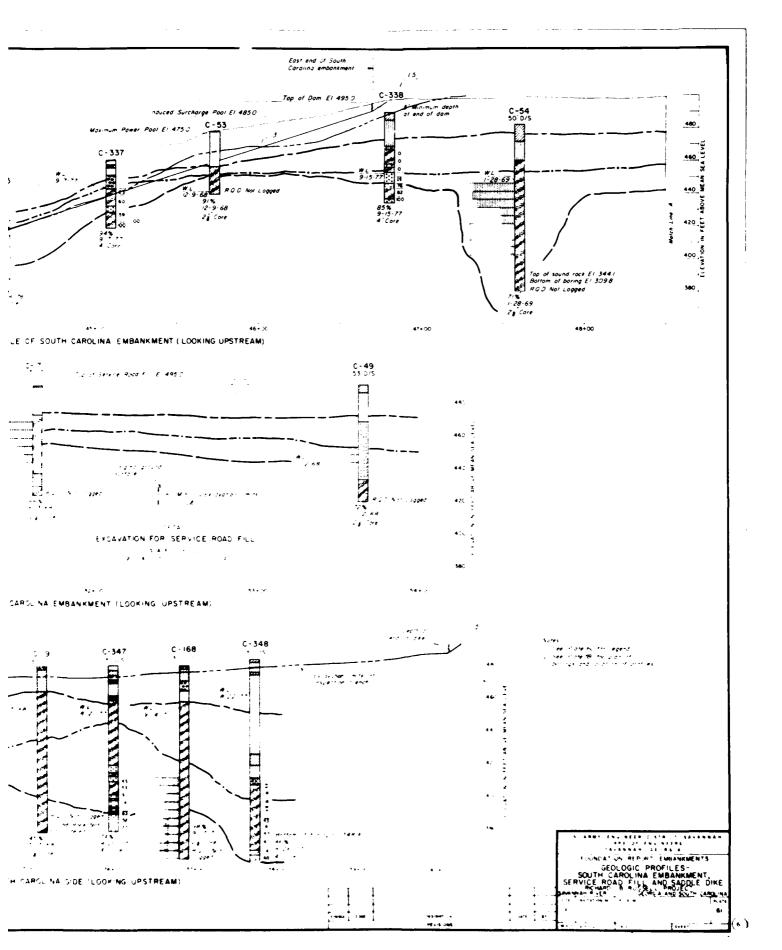
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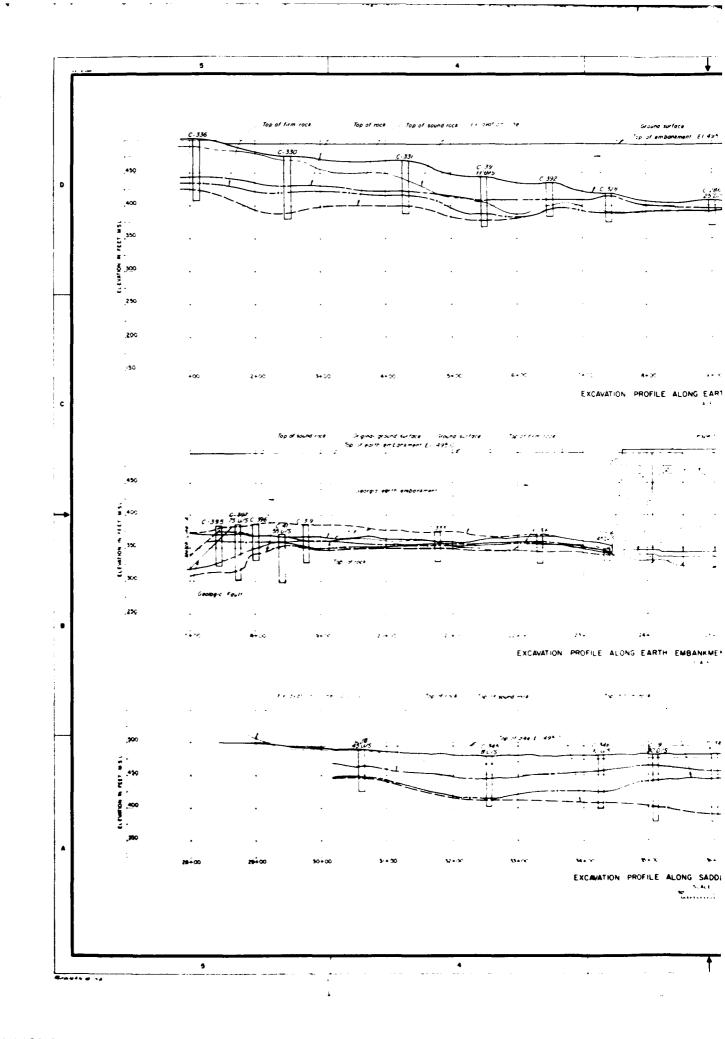
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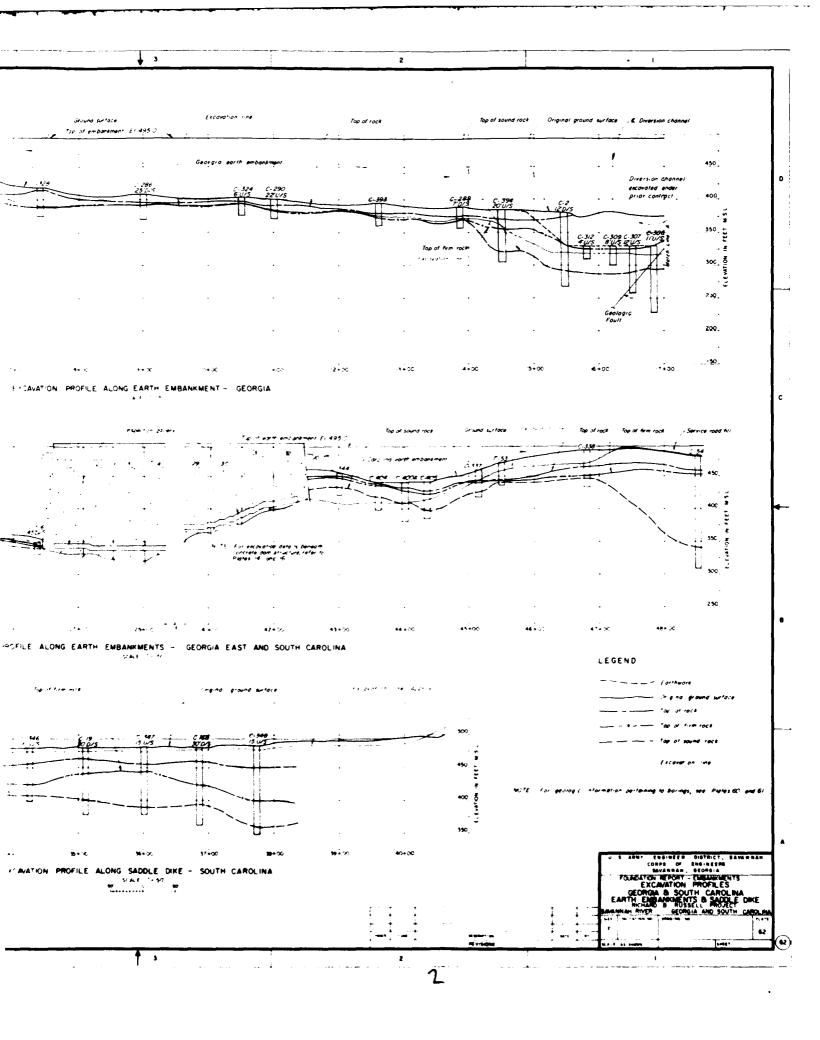


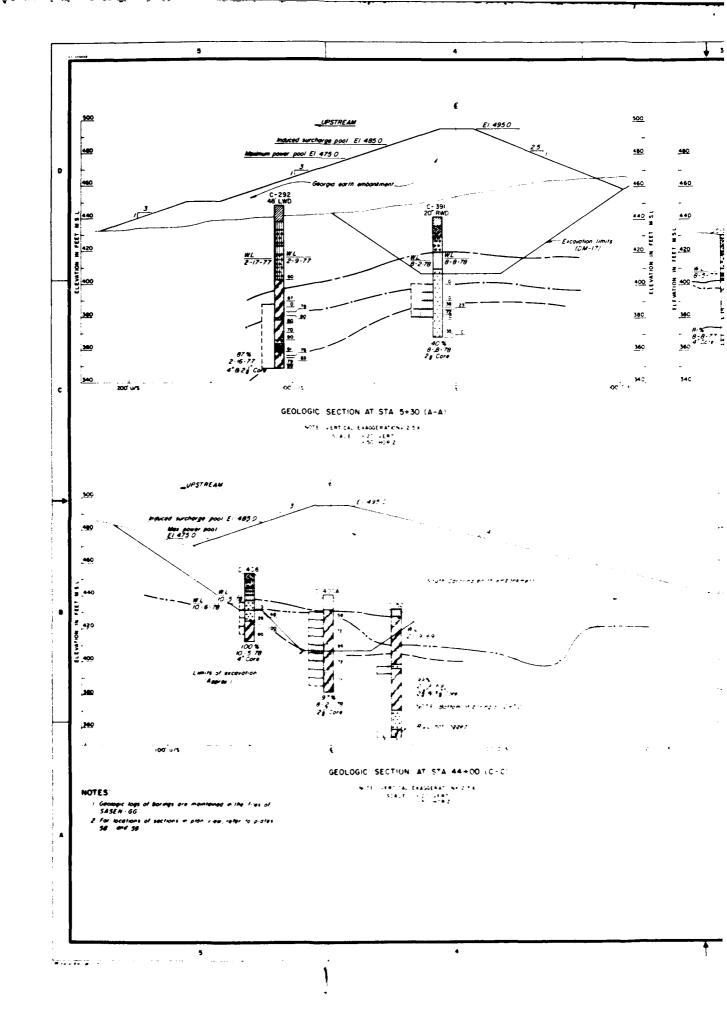


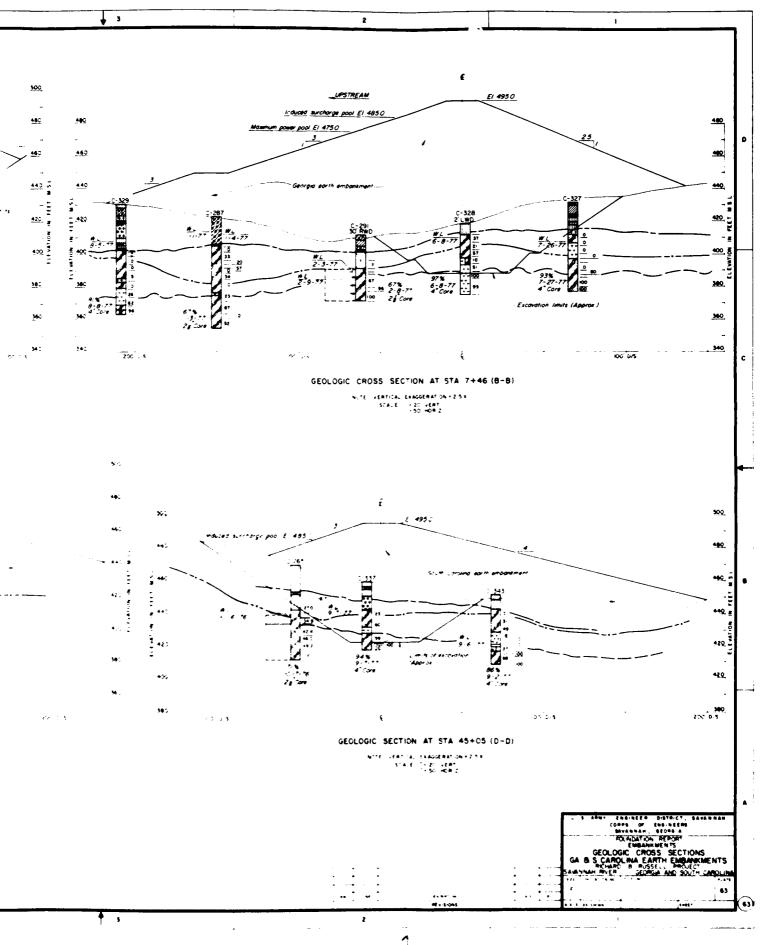


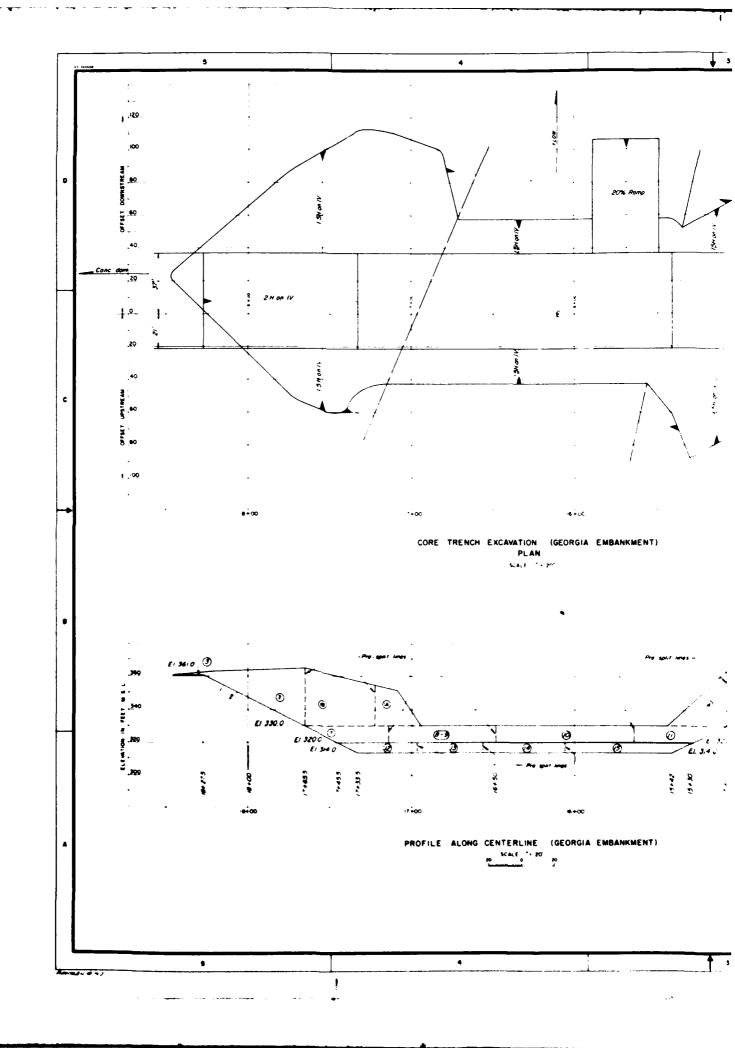


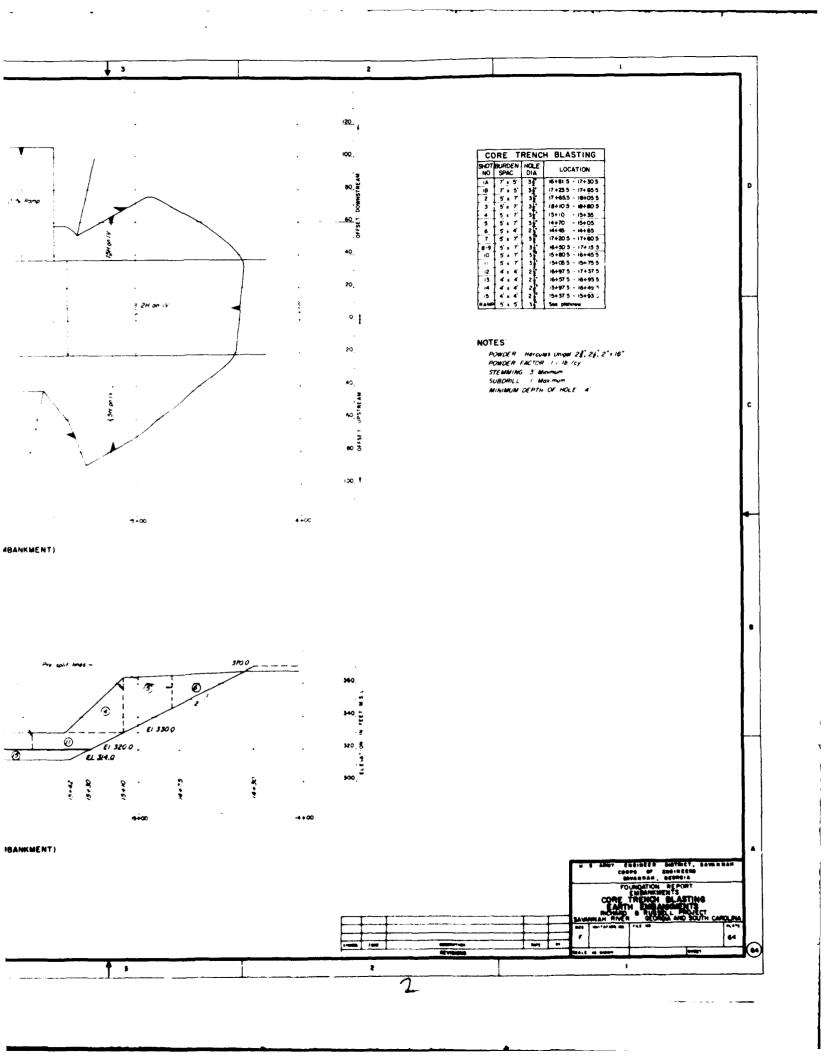




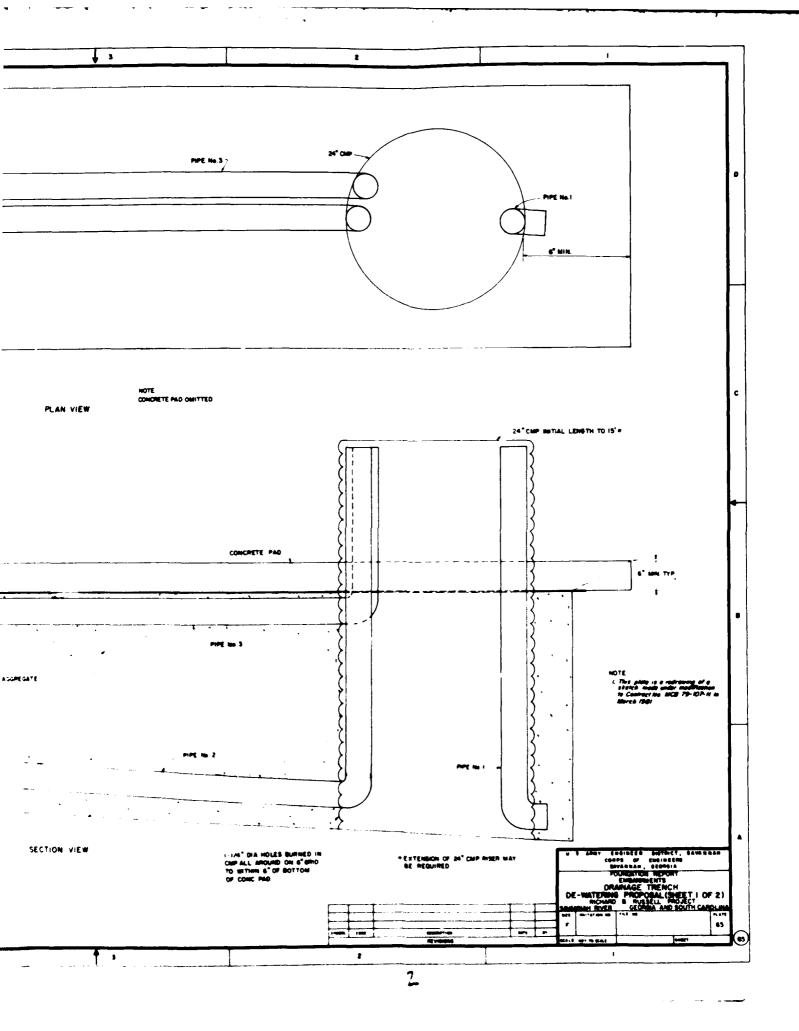


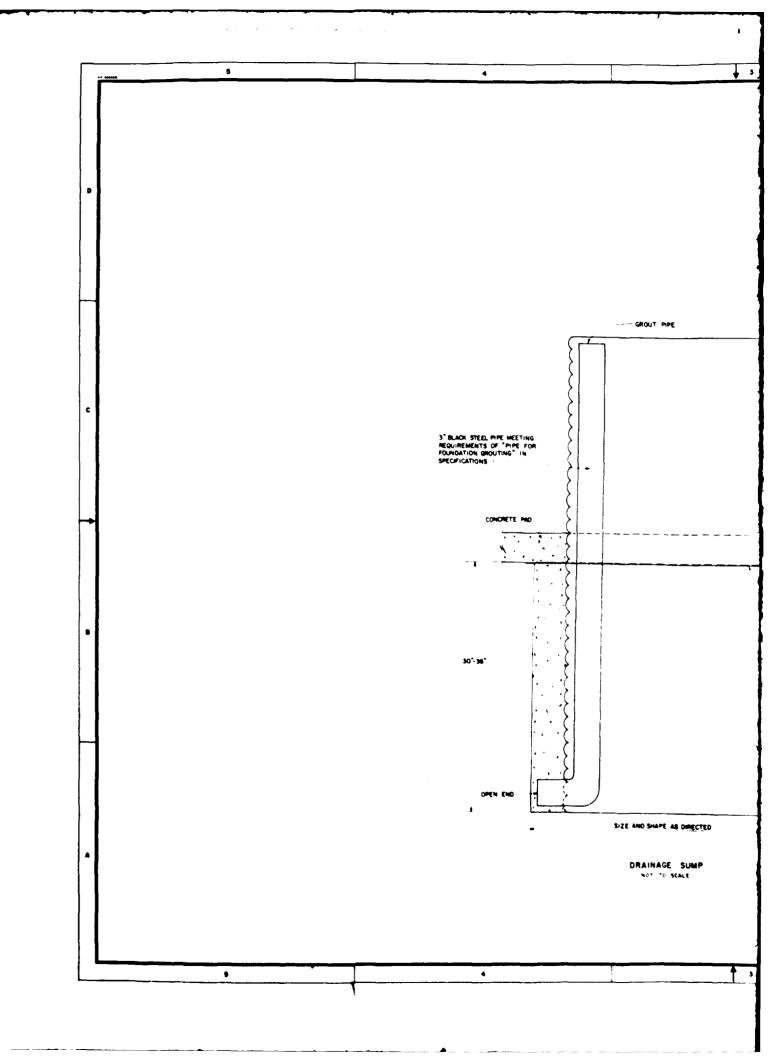


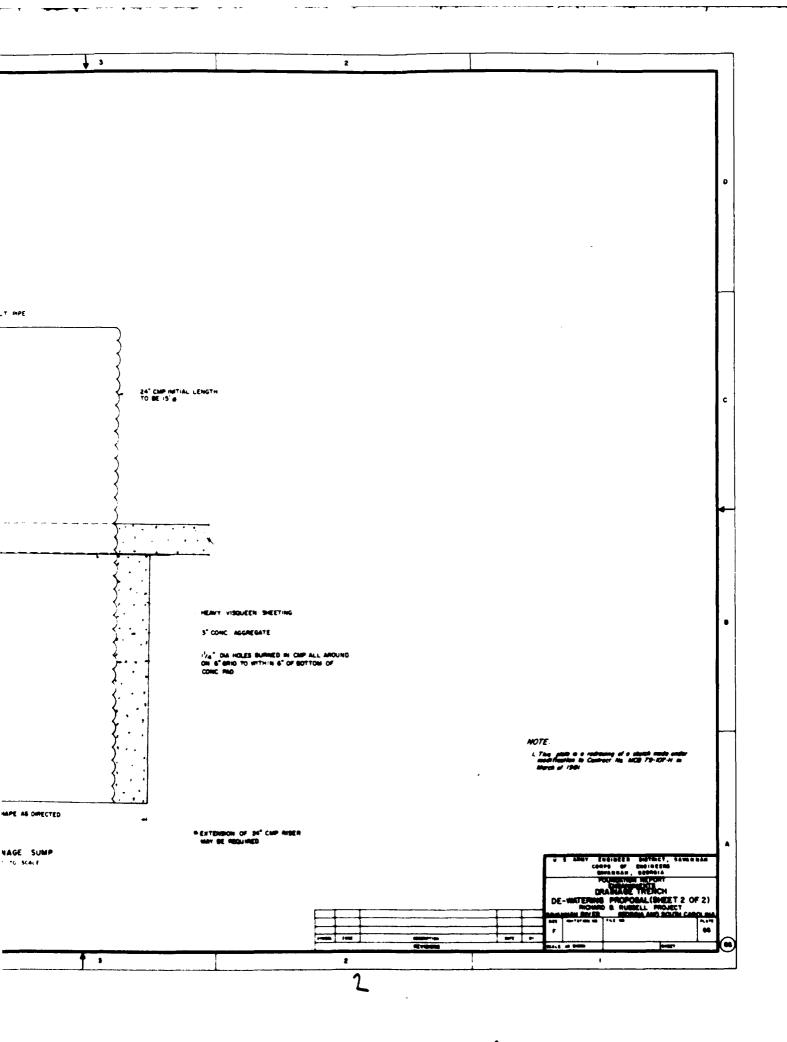


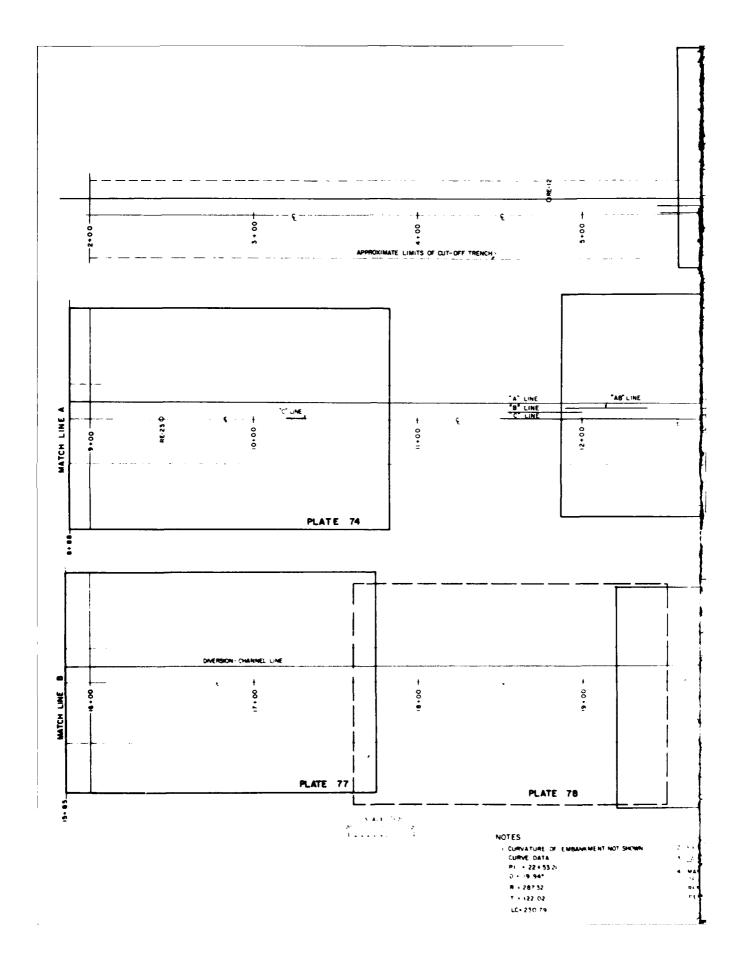


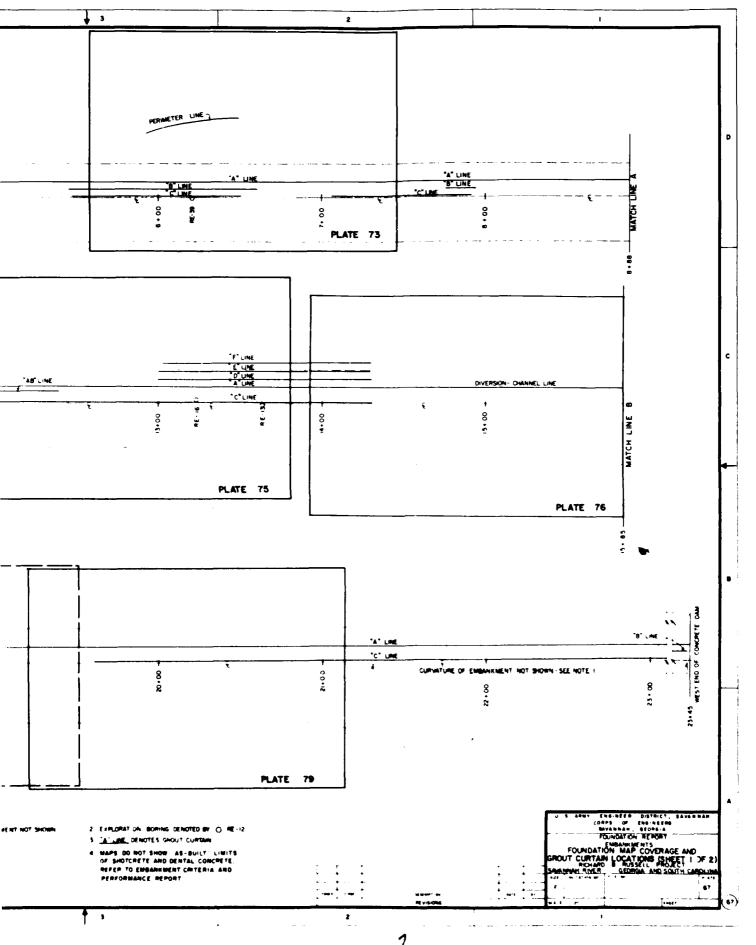
8 3'-0" LATERAL TRENCH PLAN VIEW HEAVY VIQUEEN SHEETING APPROVED POUNDATION 3" CONCRETE AGGREGATE 2 % SLOPE __ 3' BLACK STEEL PIPE, SLOTTED OR PERFORATED, AS DIRECTED FOR HALF HORIZONTAL LEWETH, PIPE SHALL MEET REQUIREMENTS OF "PIPE FOR POUNDATION SHOUTING" IN SPECIFICATIONS SECTION VIEW •



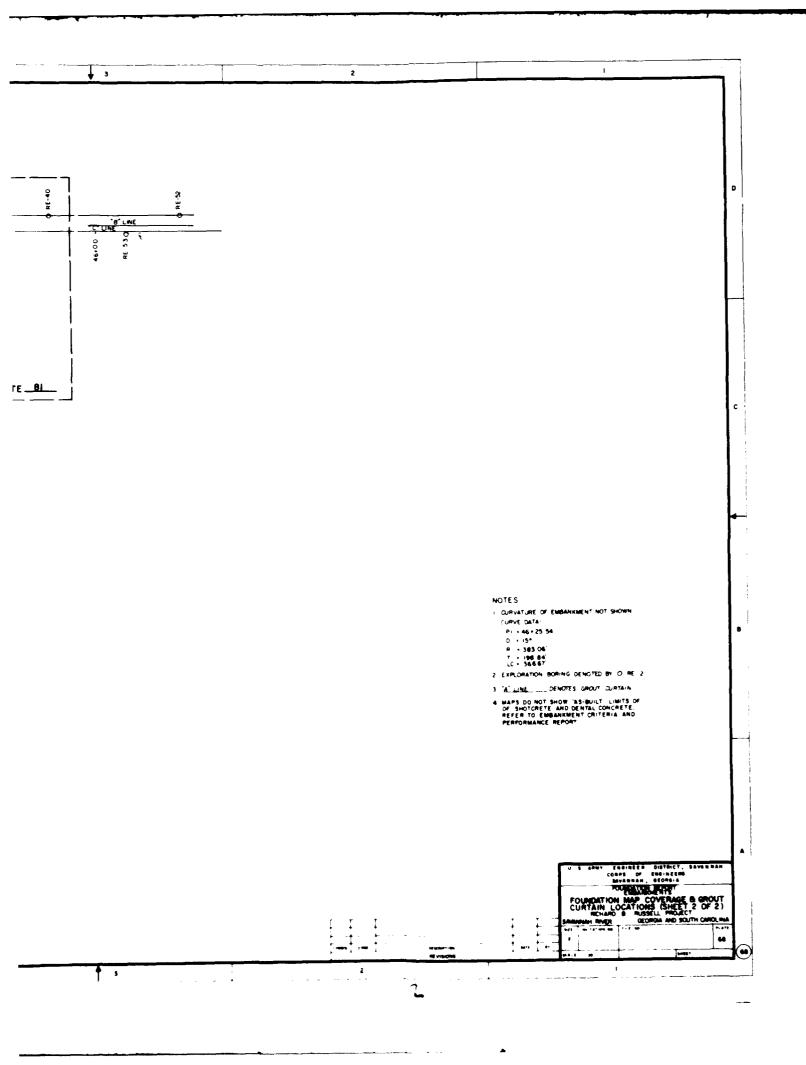


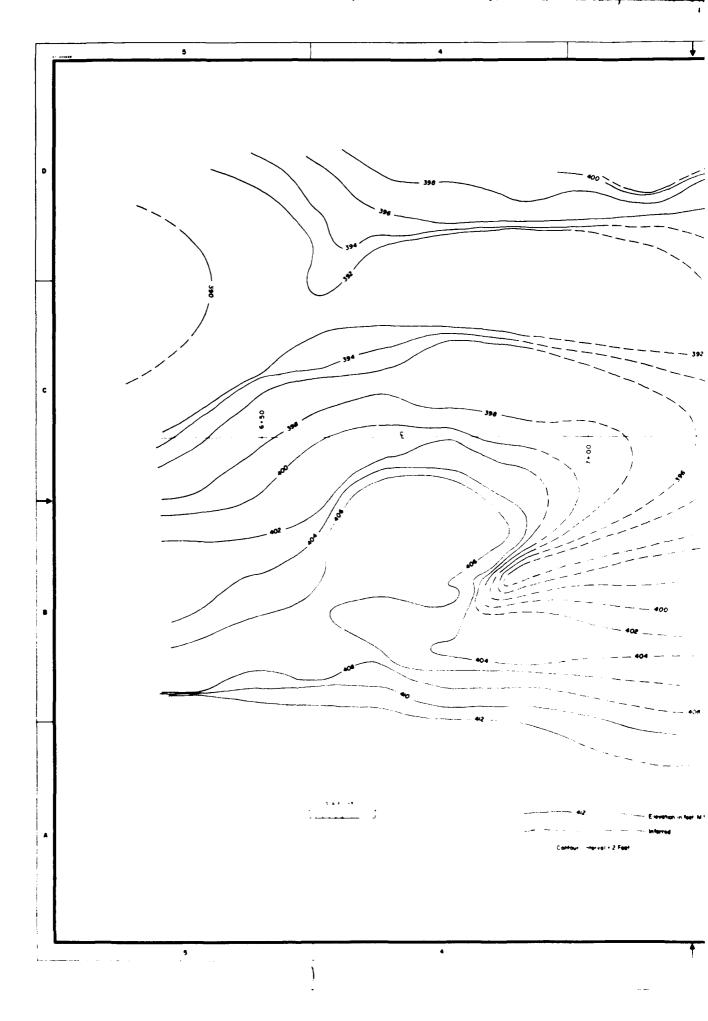


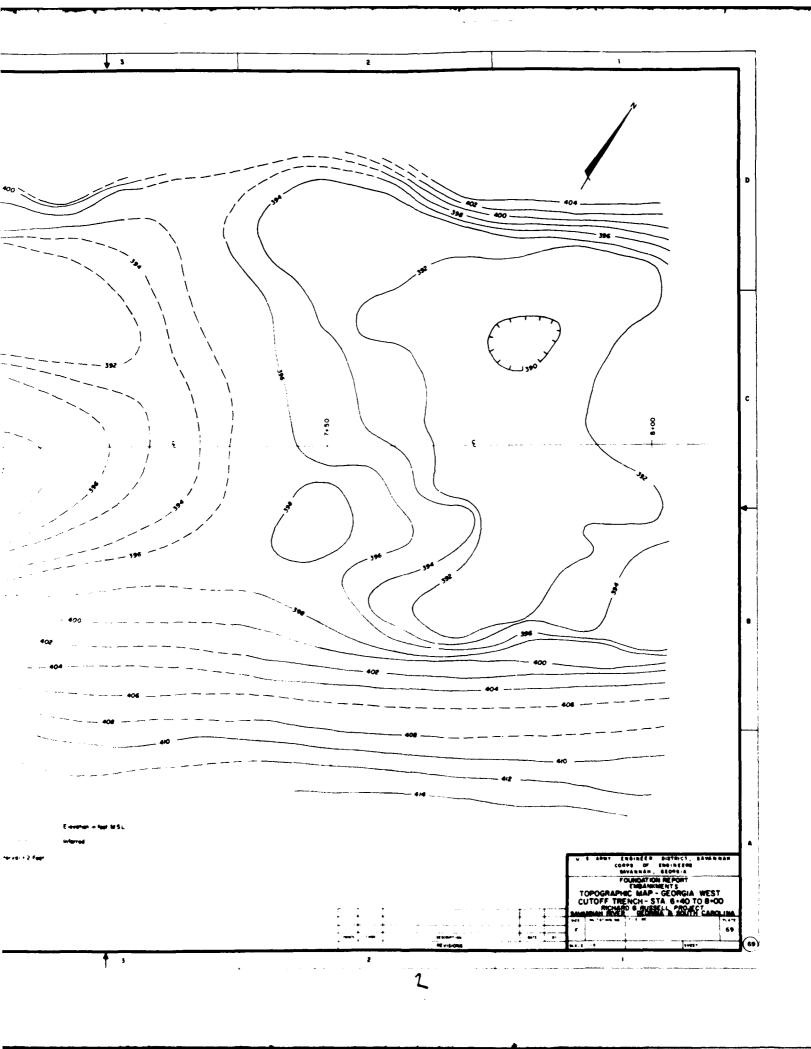




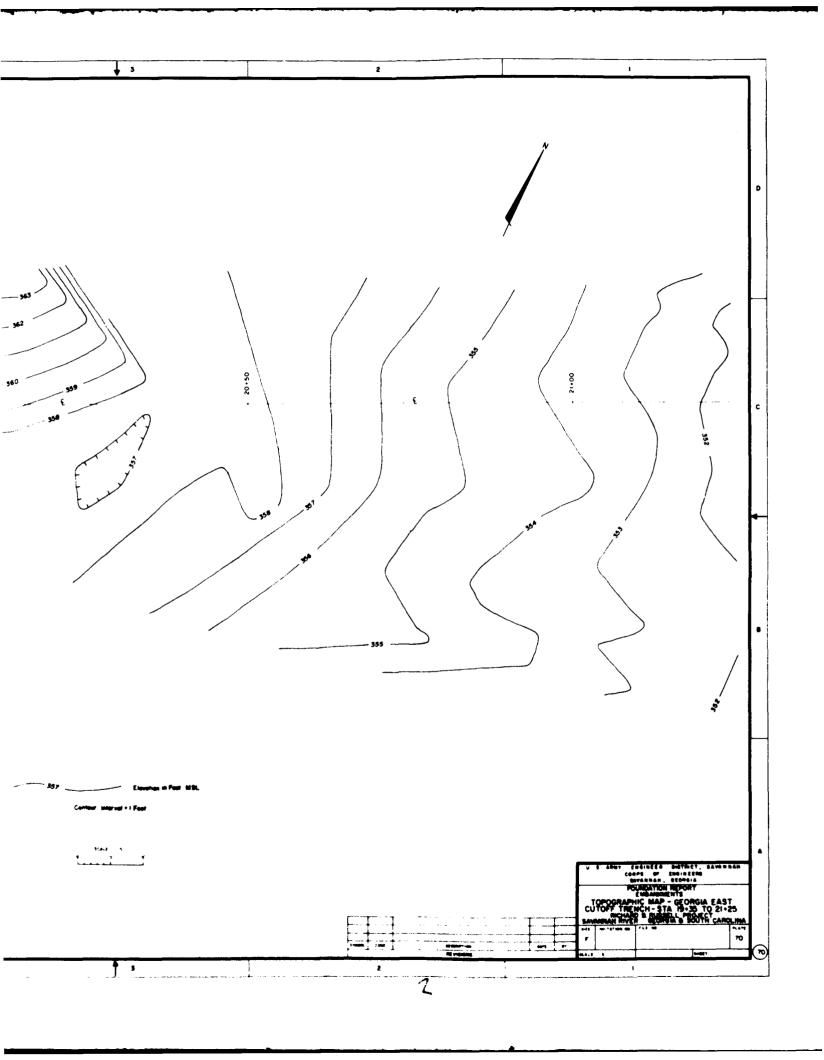
00+44 ŧ 45.00 PLATE 80 PLATE __81_ 1

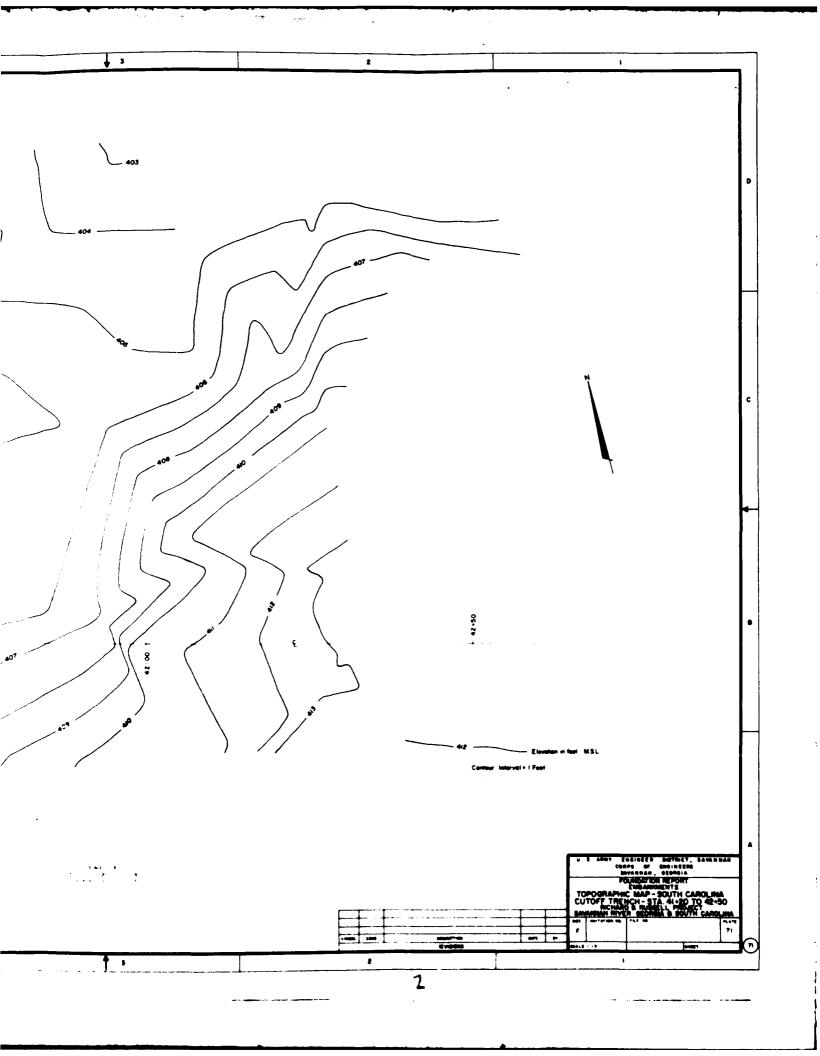




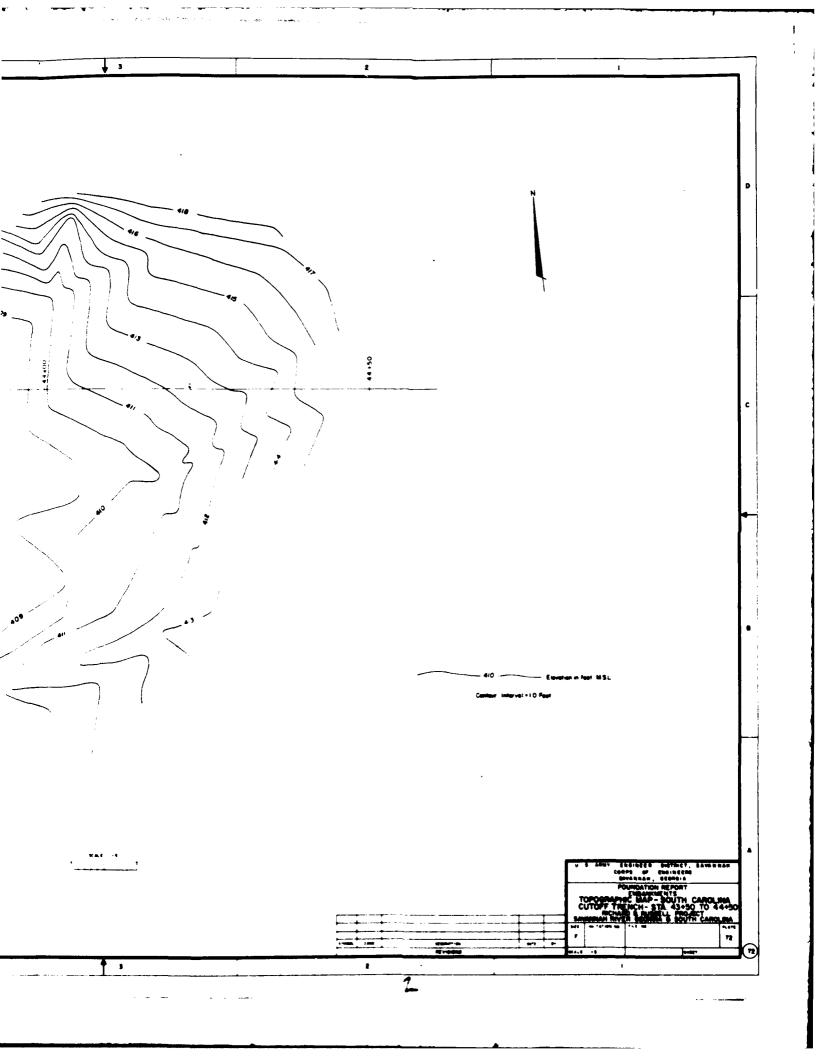


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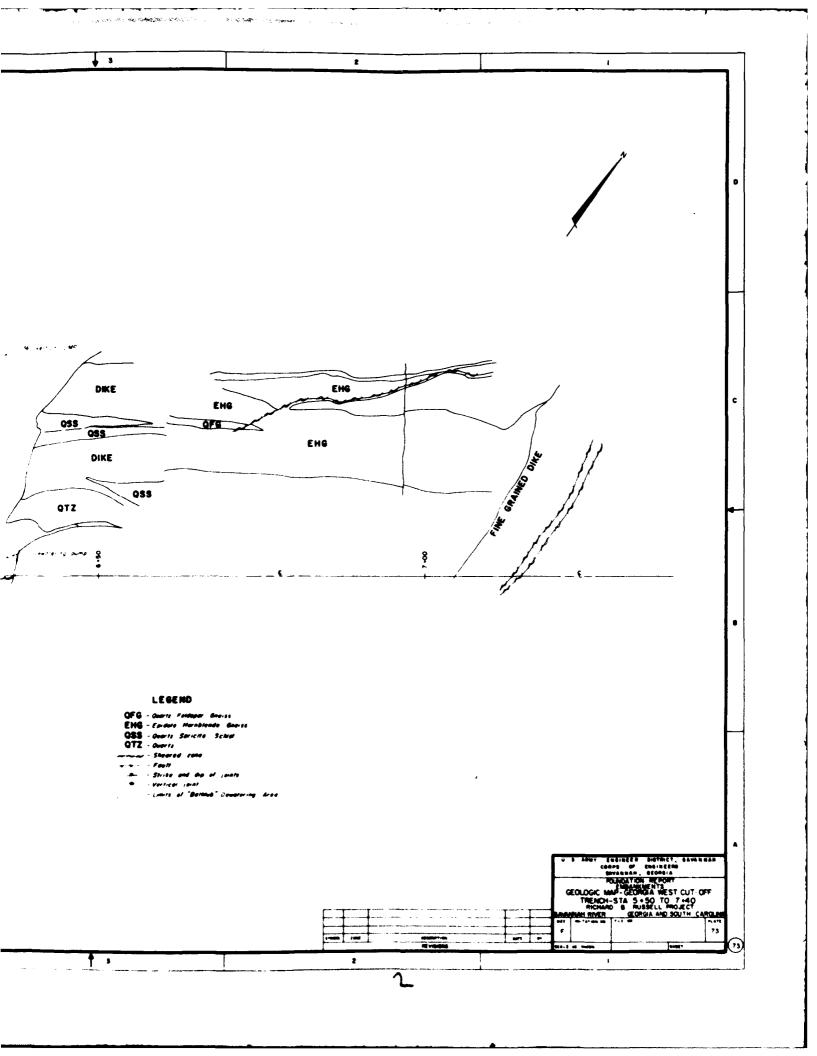


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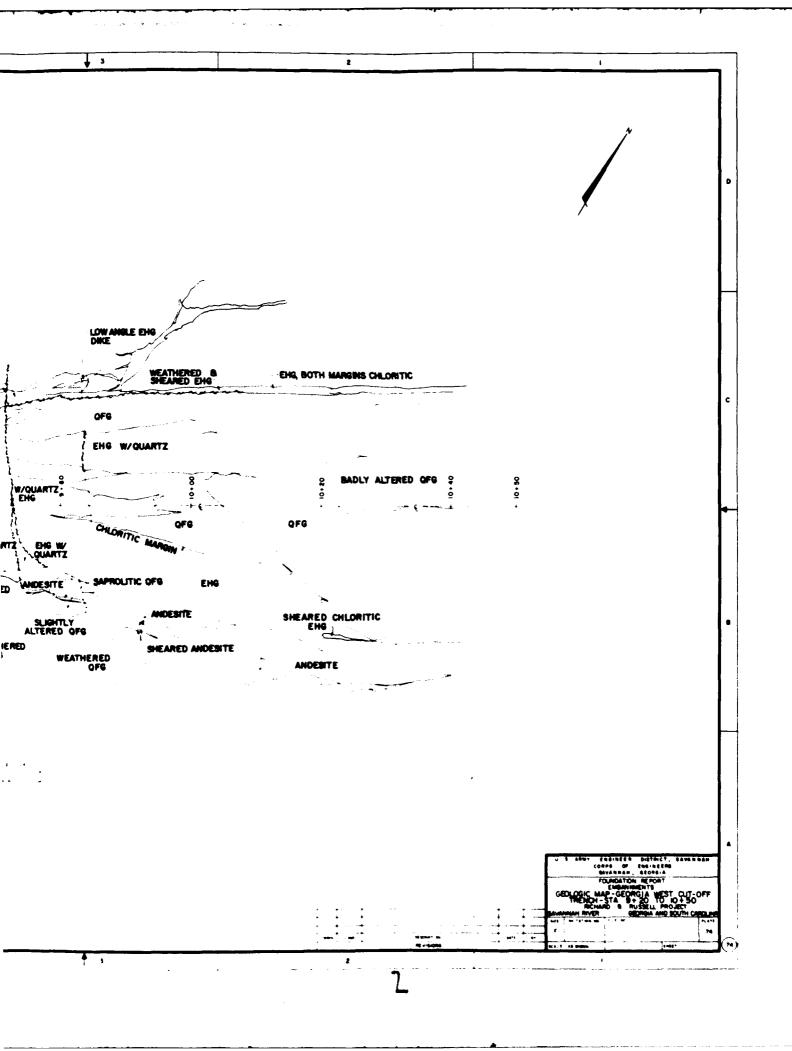


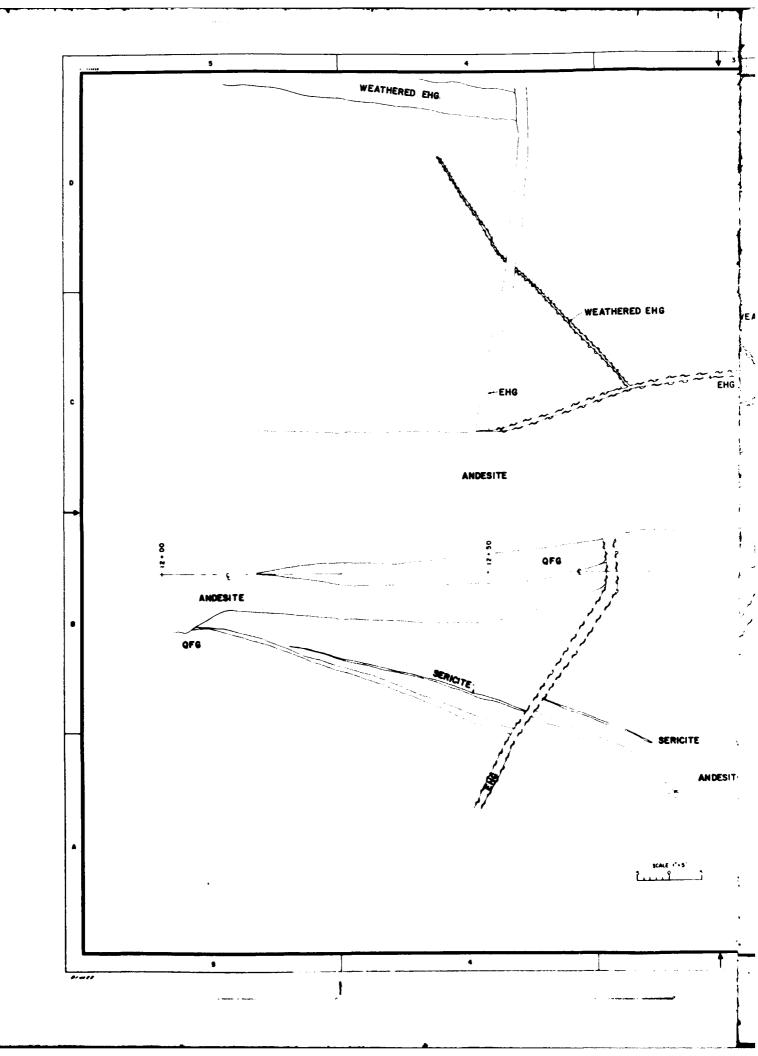
5 MOTE See Section 73-EE of text for description foundation treatment within this area DIK QSS QSS DIKE SHEARED QFG FVC drainage pizes and grave: I fer QTZ DIKE 5'ALE '-5 5 0 5 •

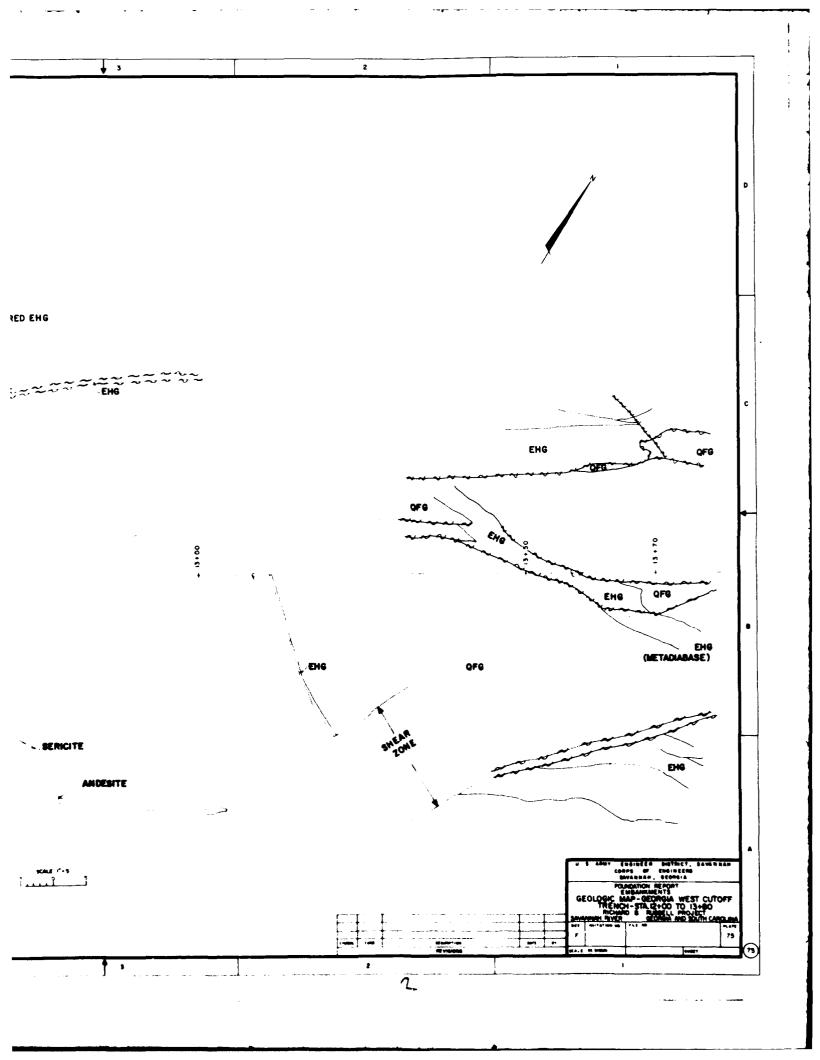
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4 LOW ANGLE QUARTZ VEIN WEATHERED SHEARED ENG EHG CHLORITIC ALONG QFG FRESH OFG MOD. WEATHERED EHG EHG W/QUARTZ SHEARED HEAVILY SHEARED ENG SLIGHTLY ALTERED JOINTED OFG EHG W/ROUNDED BLUE QUARTZ PHENOCRYSTS SLIGHTLY ALTERED OFG WEATHERED QFG WEATHERED WEATHER 514,4 (5



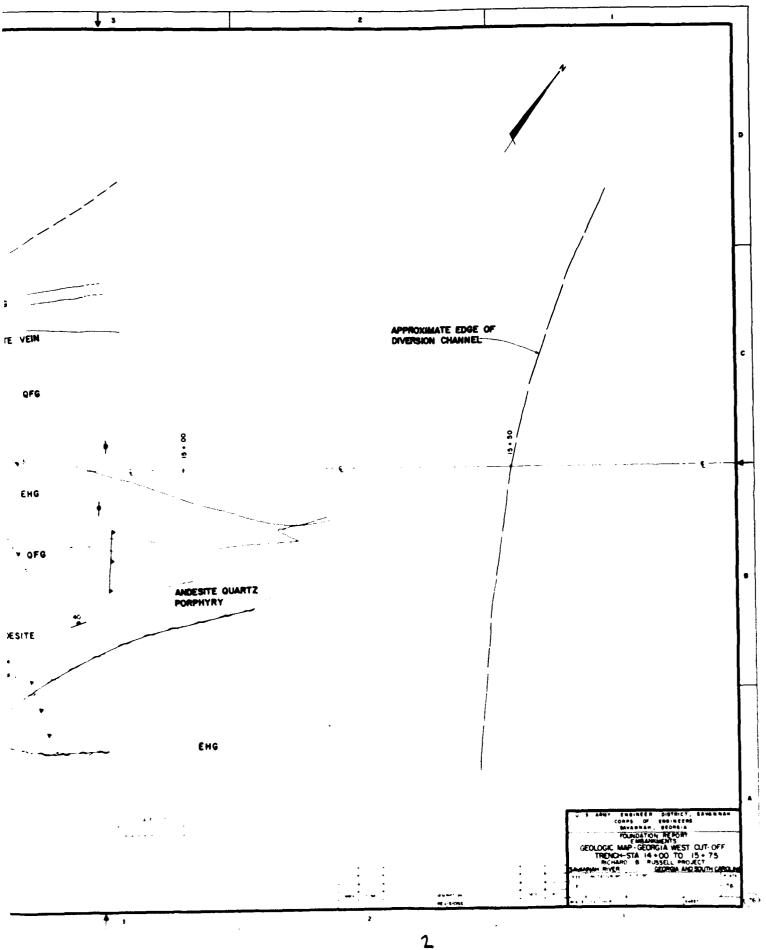


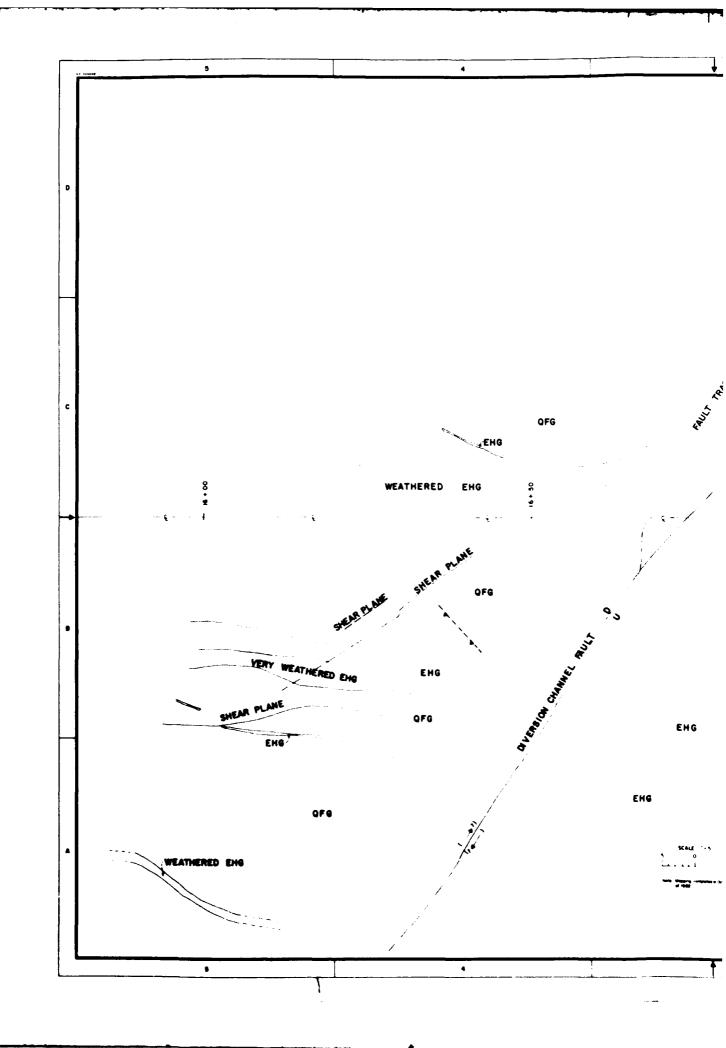


APPROXIMATE TOE OF SAPROLITE SLOPE QFG EPIDOTE VEIN QFG EHG QFG EHG EHG SHEARED QFG 10 10 ANDESITE QFG F32 ANDESITE QFG APPROXIMATE TOE OF SAPROLITE SLOPE

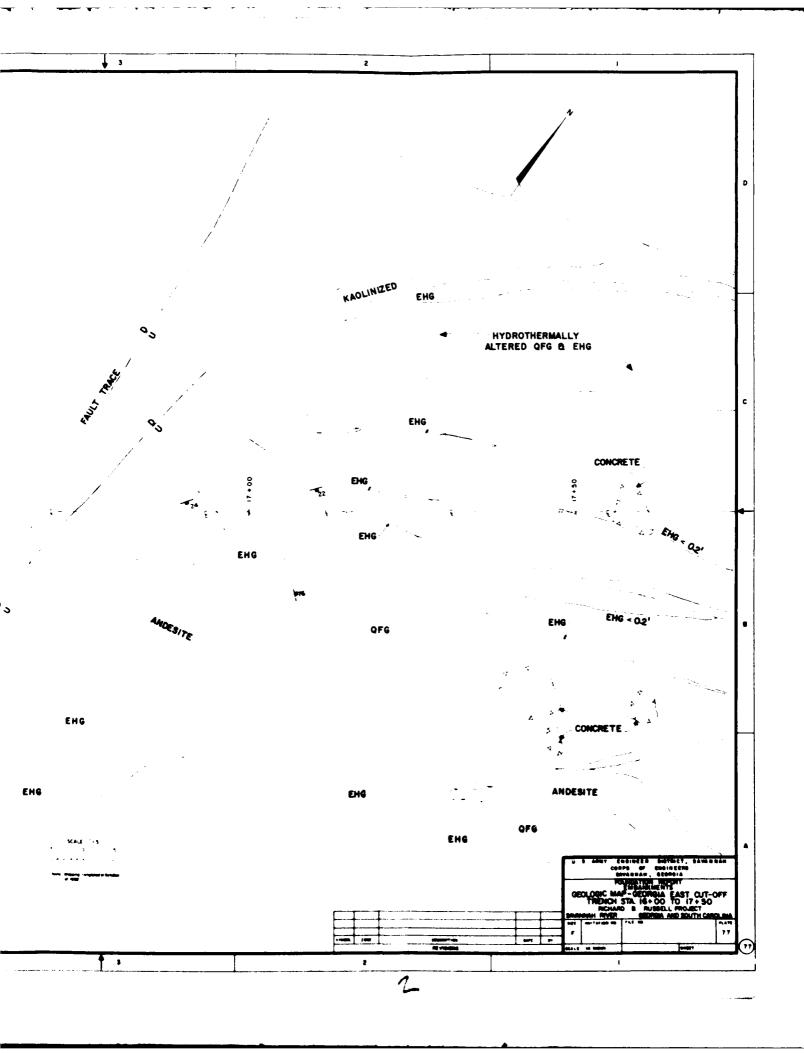
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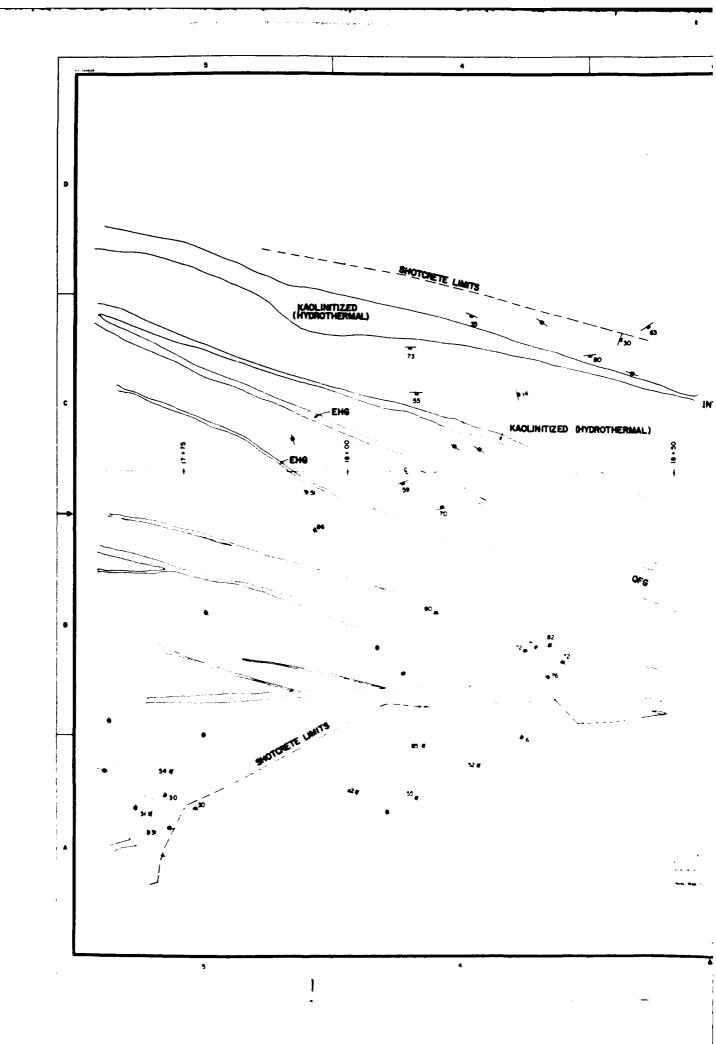
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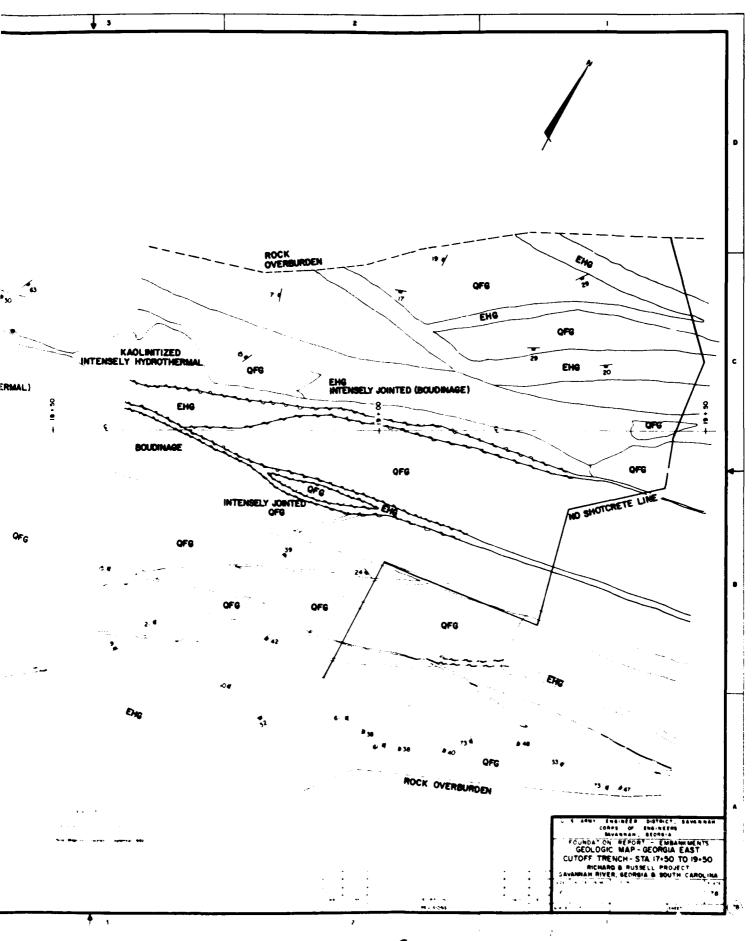


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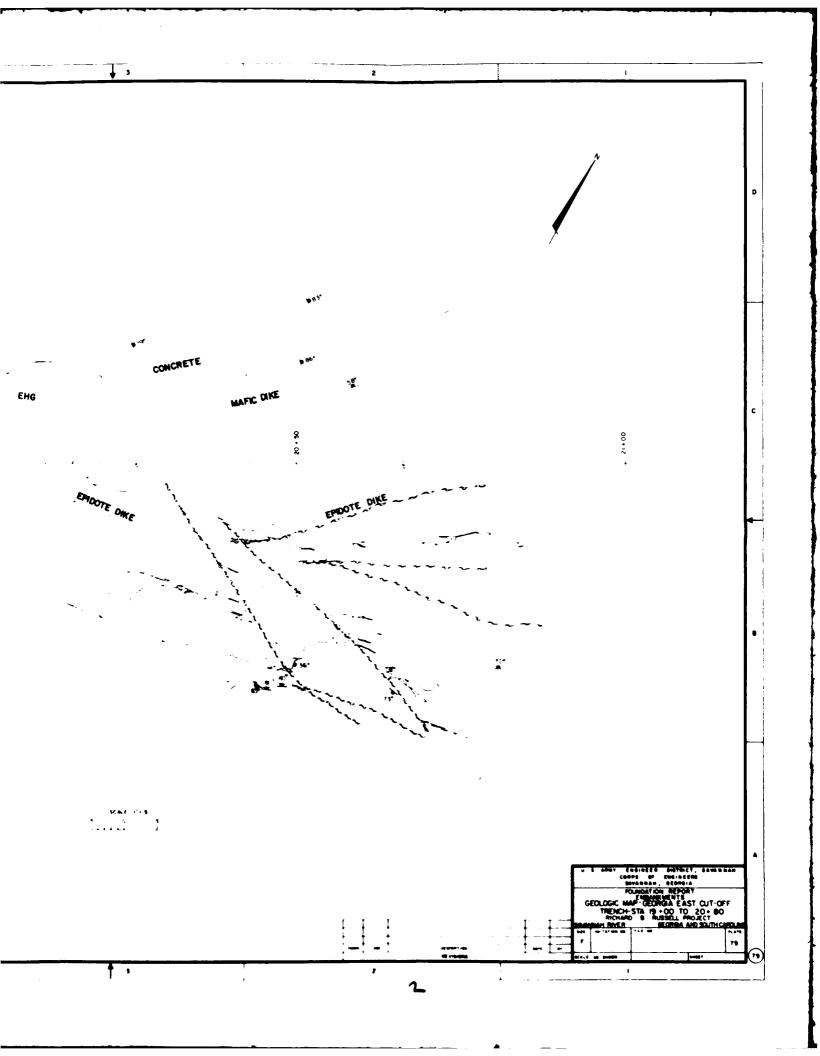




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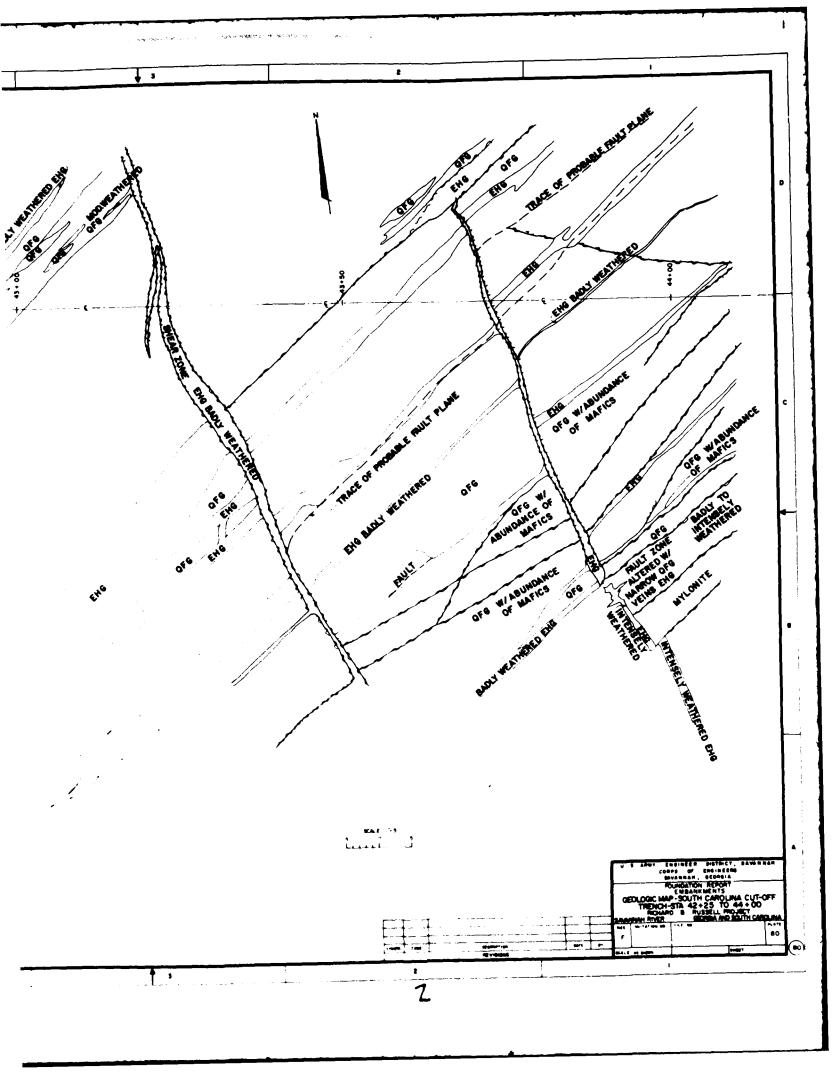


5 4 EHG EMOOTE OIK EHG SECONDARY SHEAR EPIDOTE, CALC., GOUGE ZEOLITE, CHLORITE

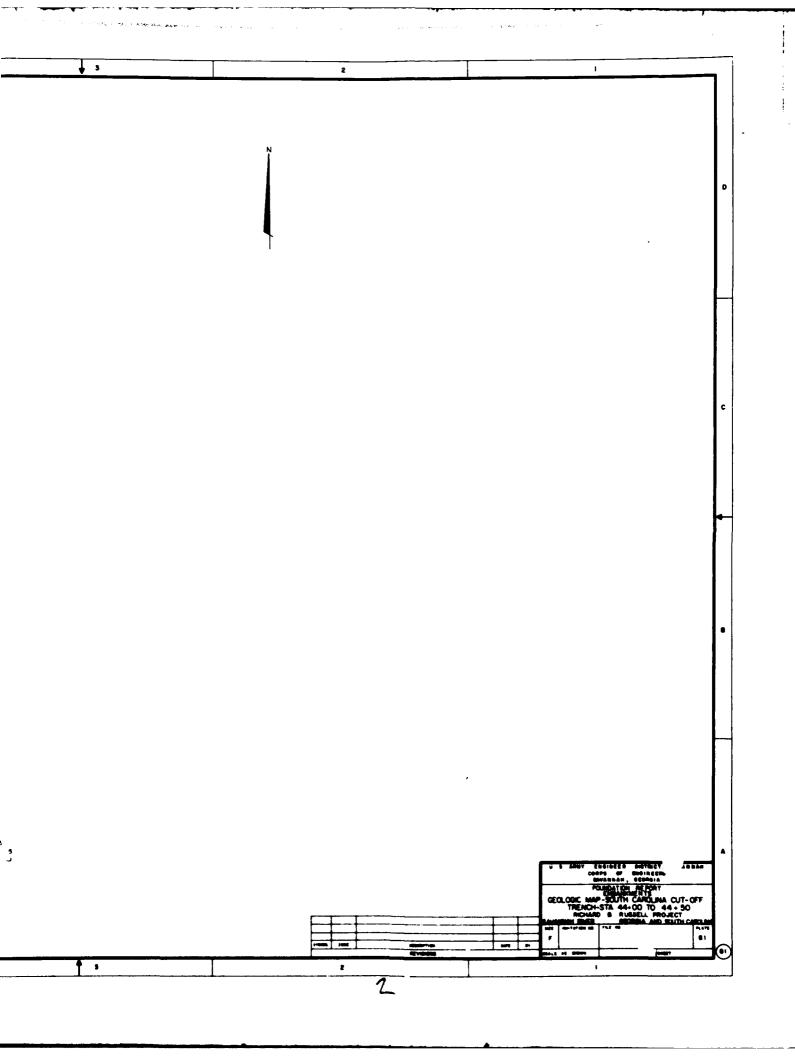


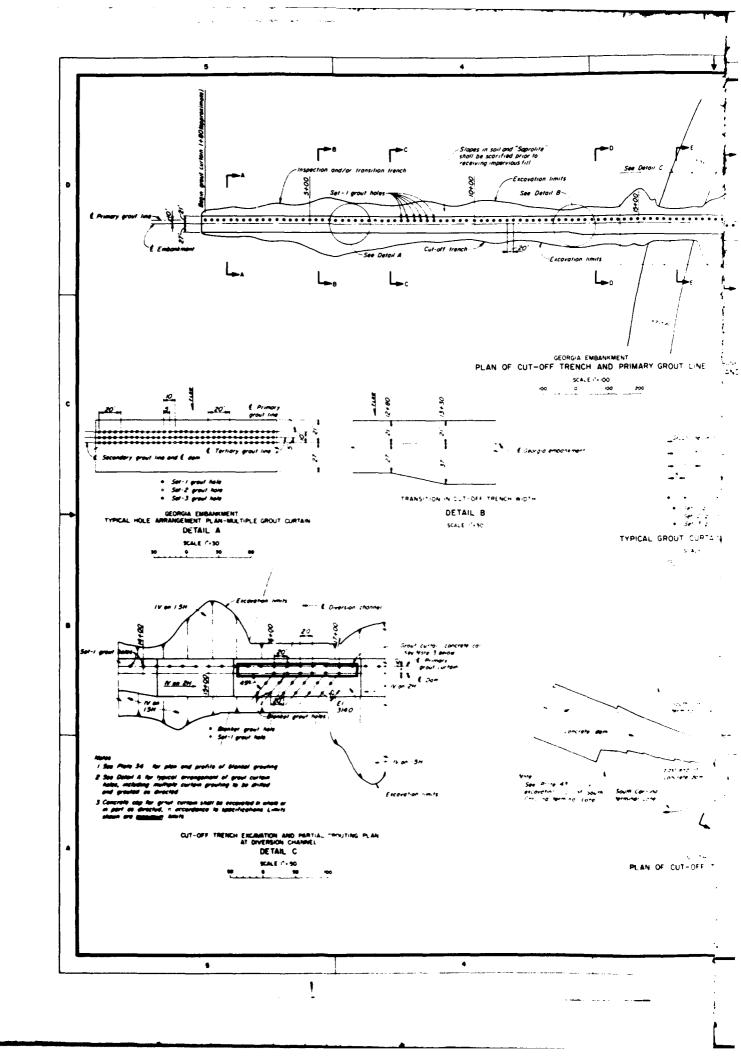
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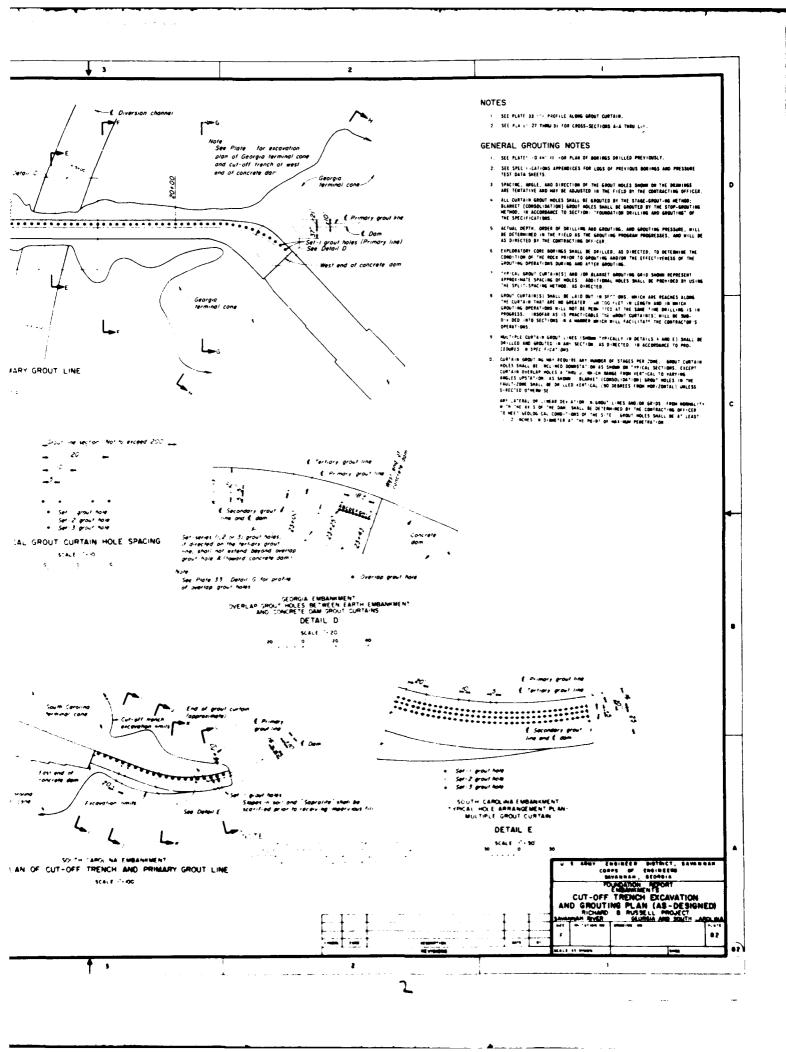
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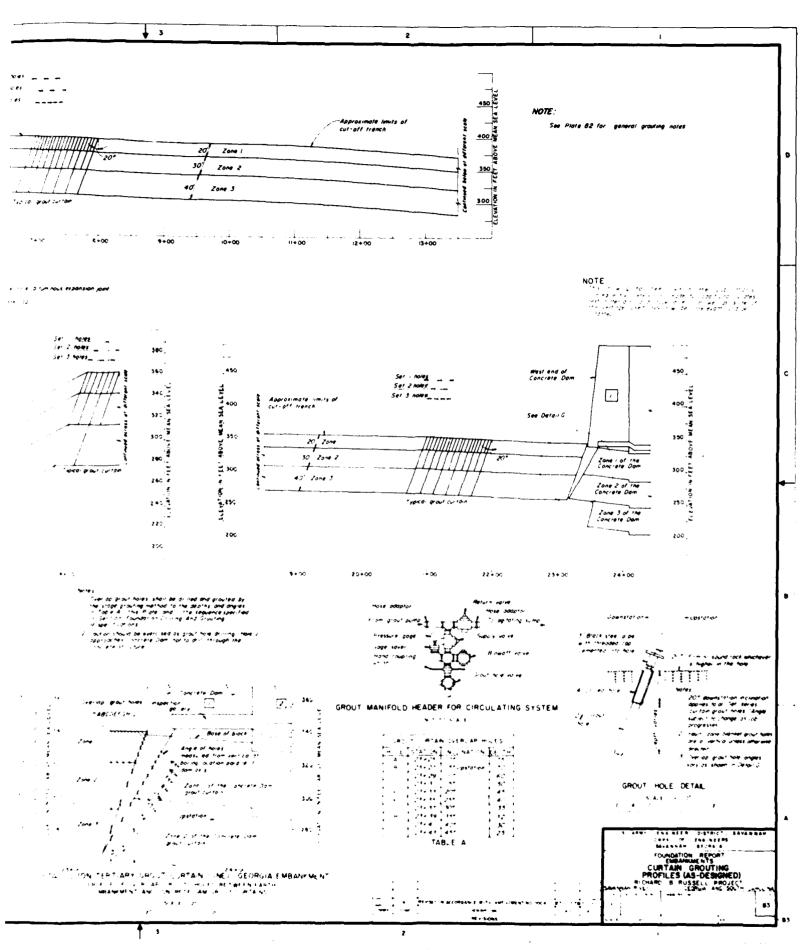


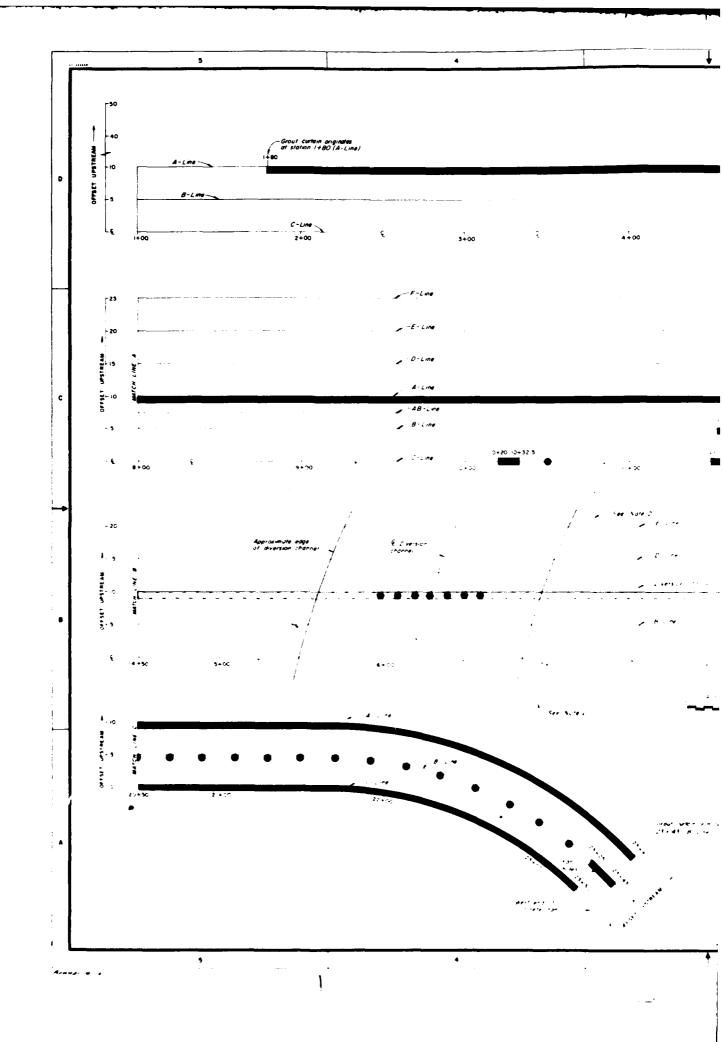
OFO EHO AND WEATHERED EN Bis out we rive to the THE WILLIAM WHILE OF WATER AUR! TOME SCALE (**5) 4 •

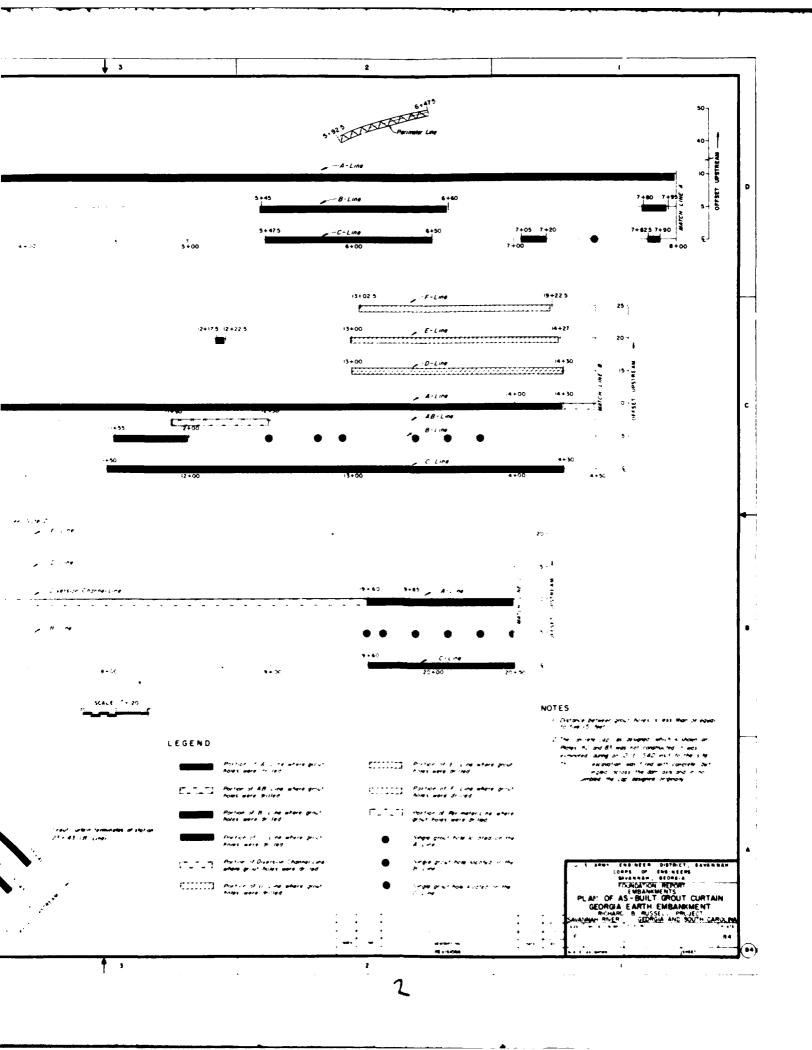






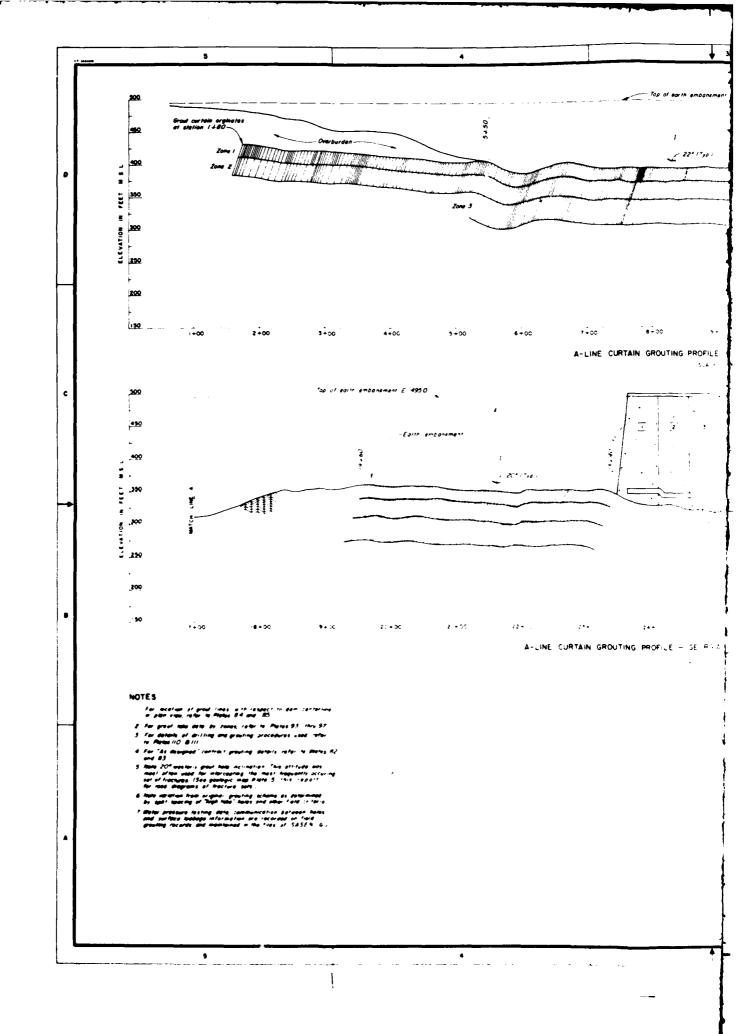


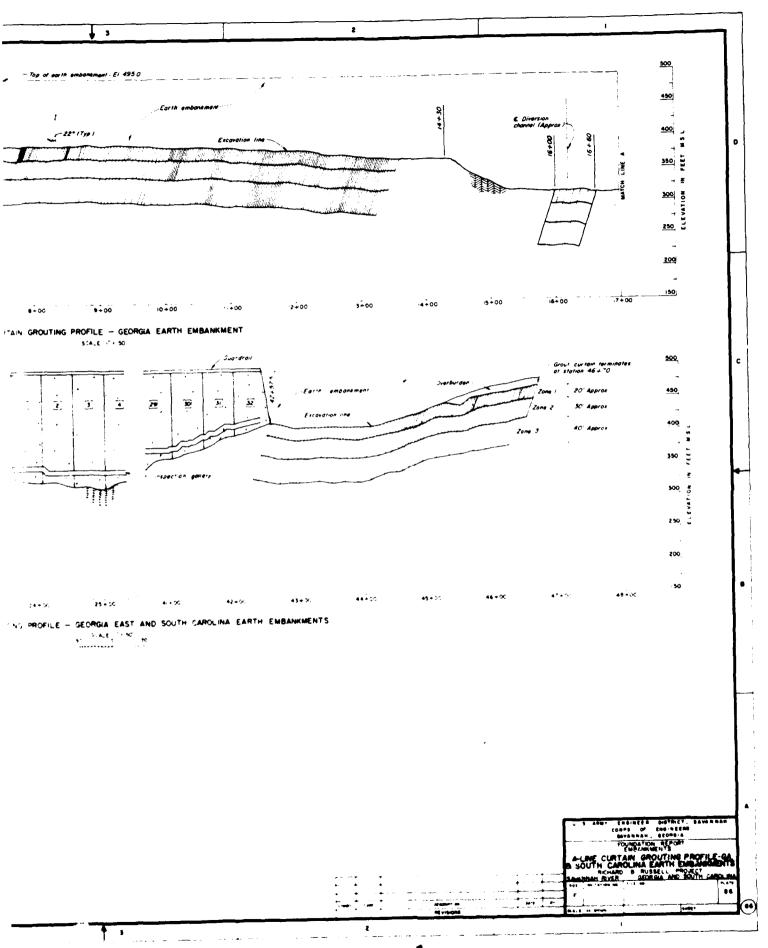




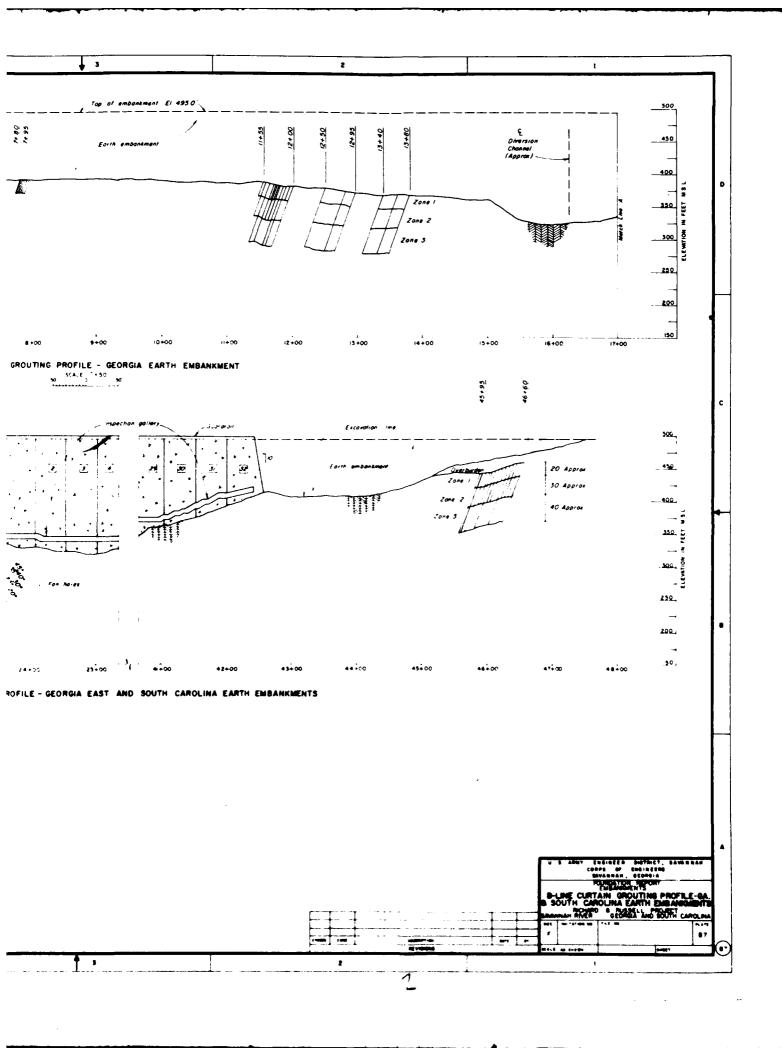
4 5 LEGEND Portion of J. Line where grout holes were drived. Cistance between grout holes is less than in equal to the 5 hear 9

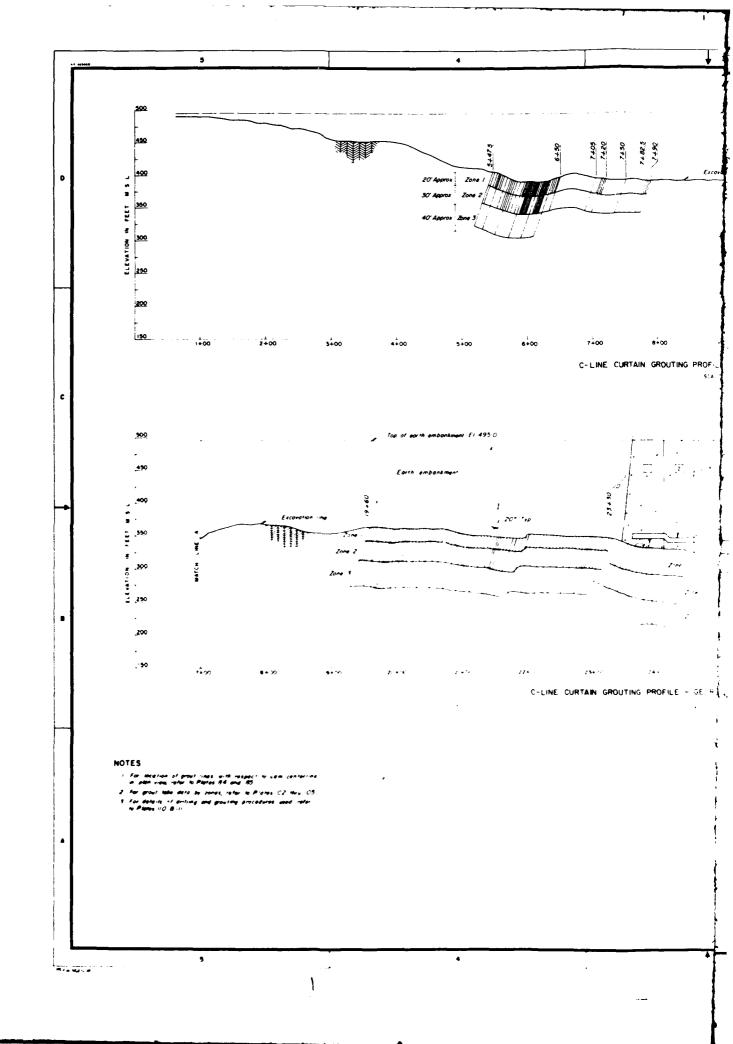
2 3 4**× Smut curtain terminates of station 46 + 75 (c), me U S ADUT ENSINEES DISTRICT, SAMESHAN CORPS OF ENGINEERS BUNGANIAN SECONDA TO PROPERTY ENGINEERS PROJECT CURTAIN SOUTH CAROLINA EARTH EMBANKMENT RICHARD B RUSSELL PROJECT LANGINGH TO B RU 1_

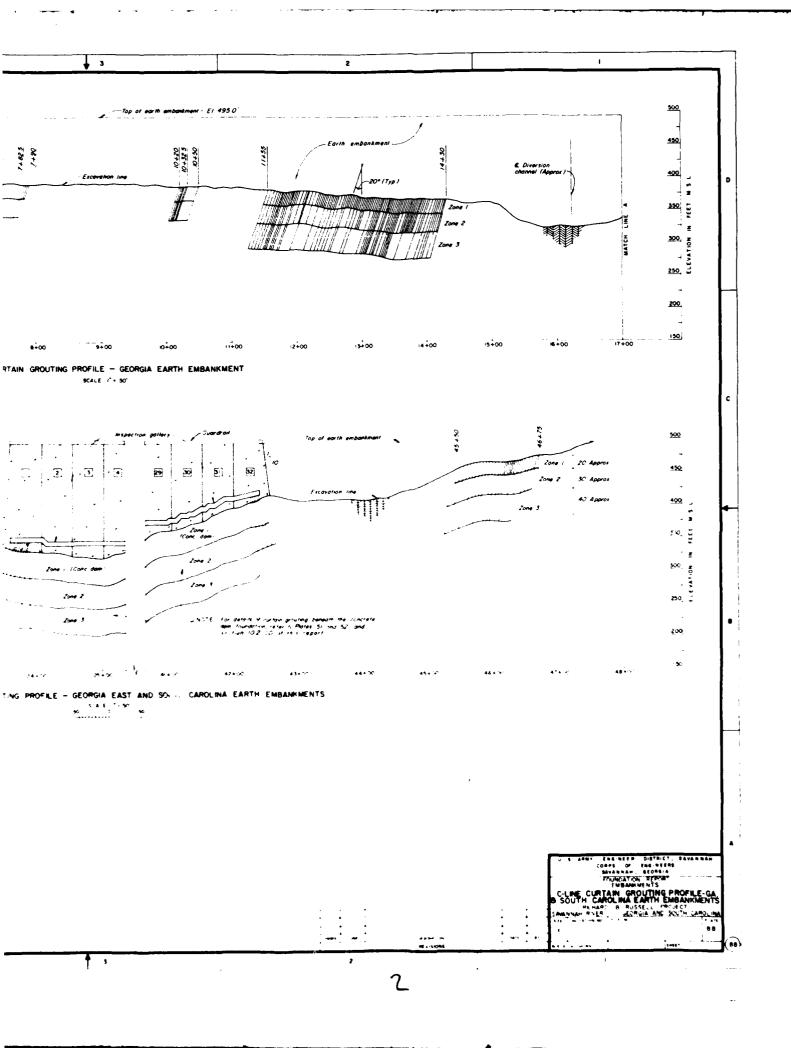




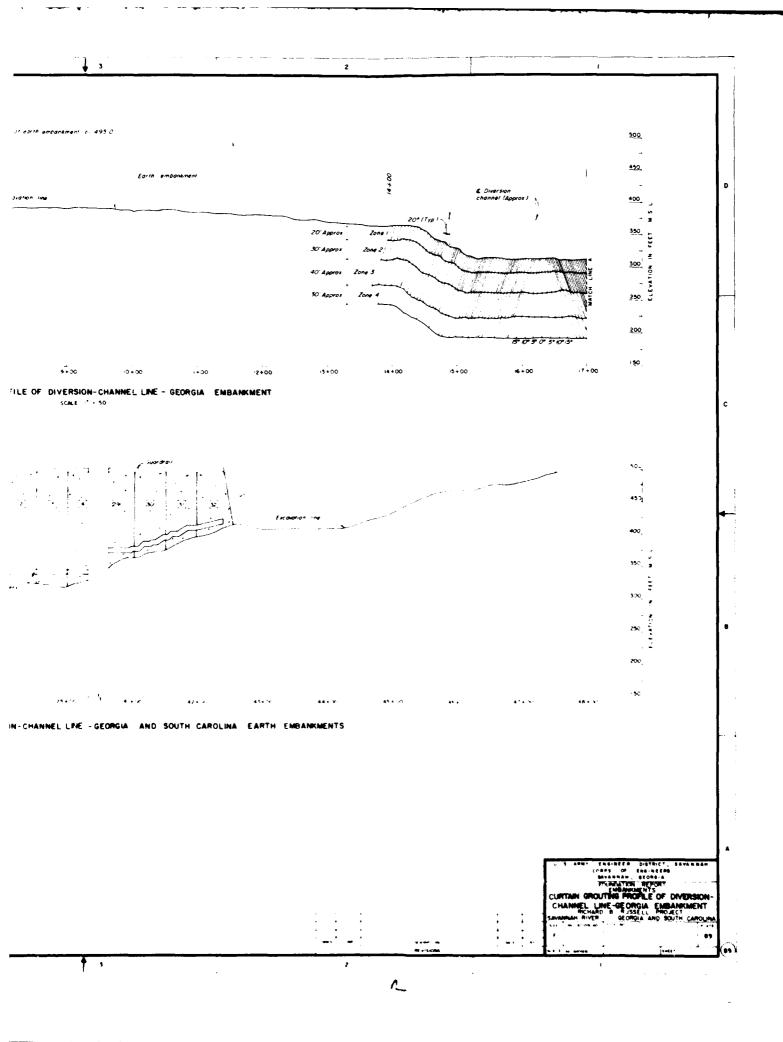
5 7 95 20° 400 20' Appros 30 Appros 300 250 **-** --200 ...50 6+00 4+00 7+00 5+00 1+00 2+00 3+00 8+00 B- LINE CURTAIN GROUTING PROFILE Top of embantment El 4950 IIII Ę.330 300 200 **عد**د. B-LINE CURTAIN GROUTING PROFILE - FORGIA ,

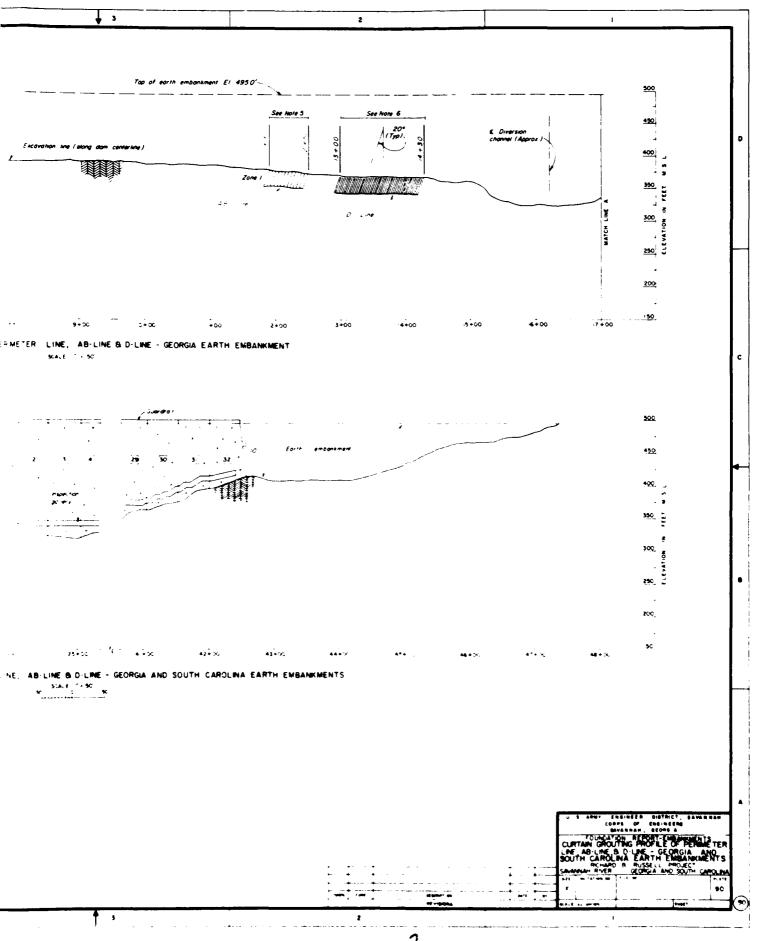


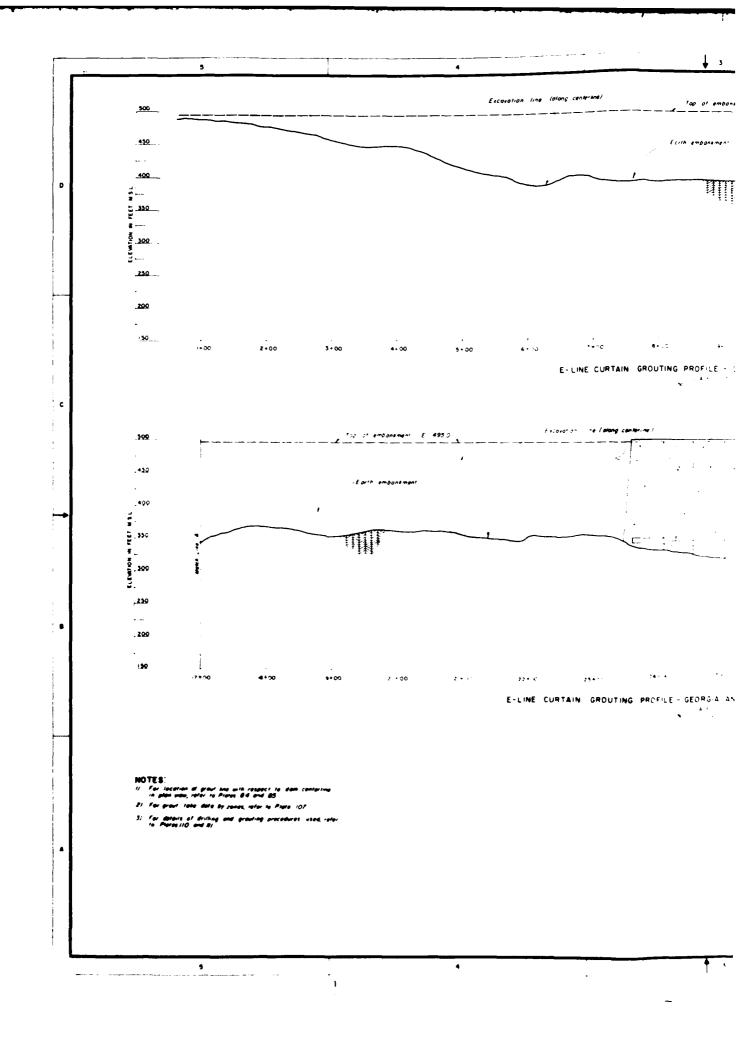


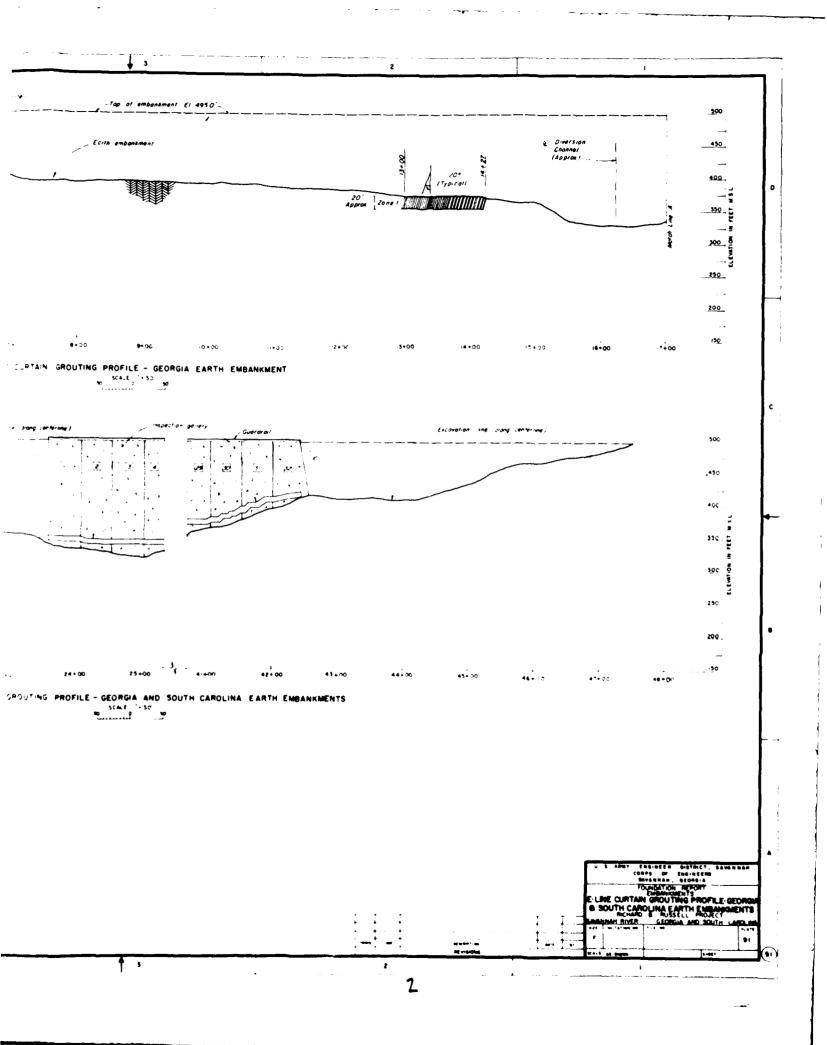


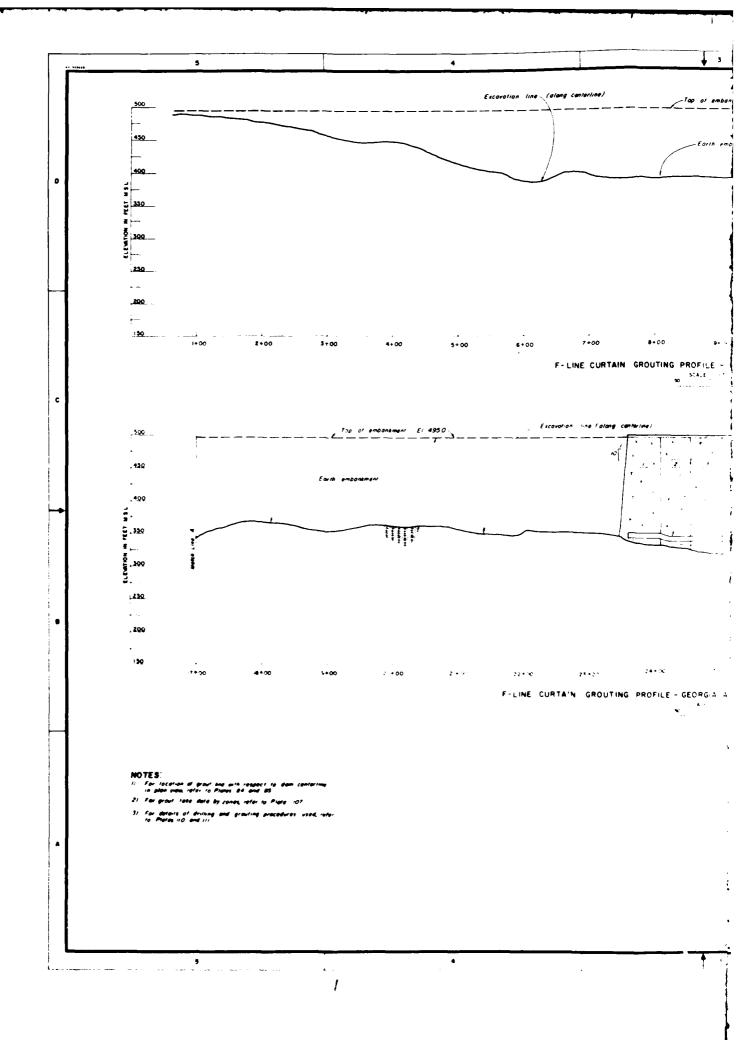
500 450 Escaration line 350 300 200 (150 _ 1+00 6+90 8+00 2+00 3+00 5.00 7+00 CURTAIN GROUTING PROFILE OF DIVERSION-,220 7.00 22 . -2 + (4. CURTAIN GROUTING PROFILE OF DIVERSION-CHANNEL LEGE 1 I. For location of growt times with respect to dam co in plan troop, robot to Photog 84 and 83. Fire growt spin date by Joines, refur to Photo 'Ob-3. For denotes of drilling and growting procedures wood, to Plane III, 8.111.

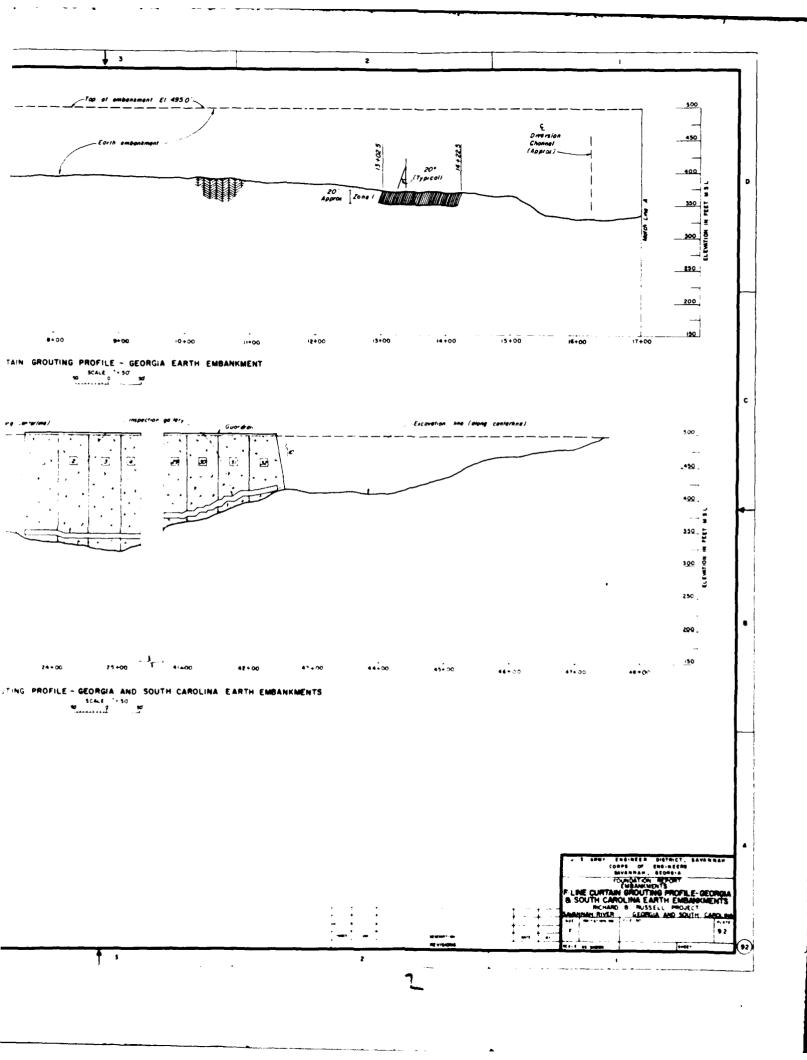


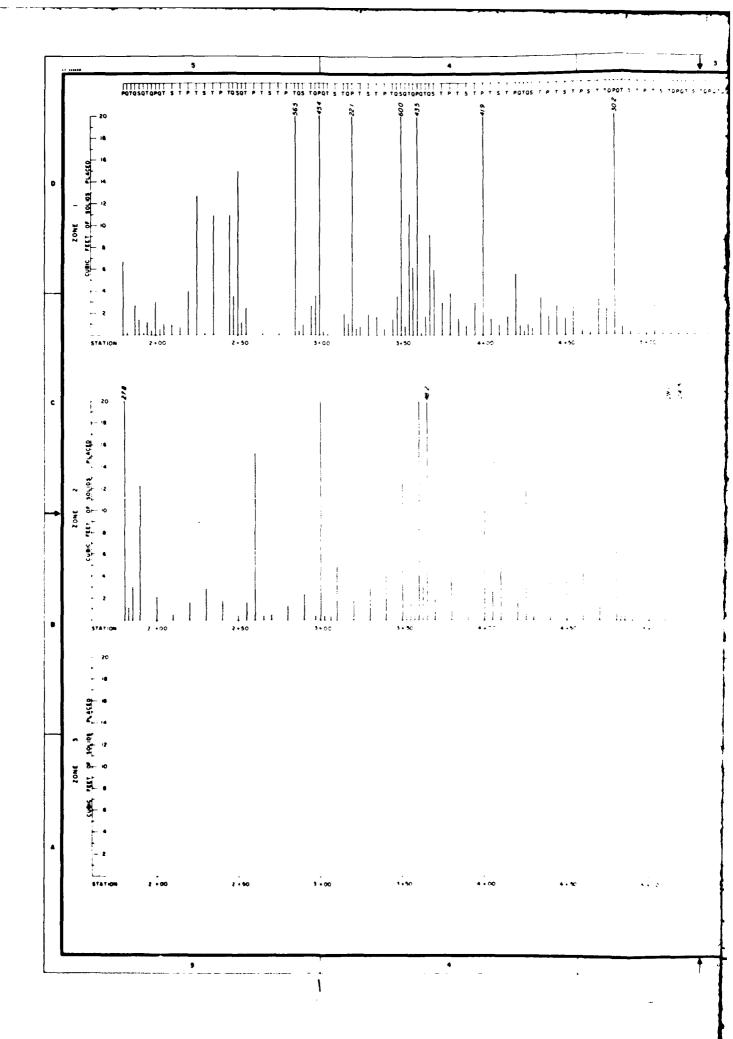


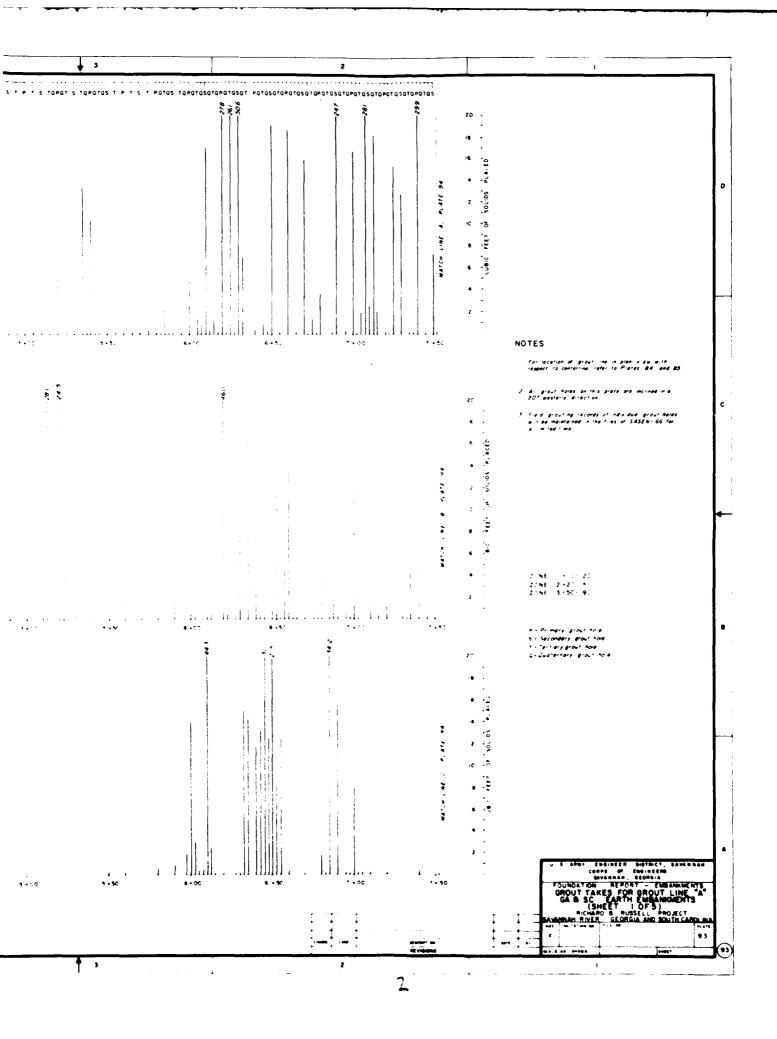


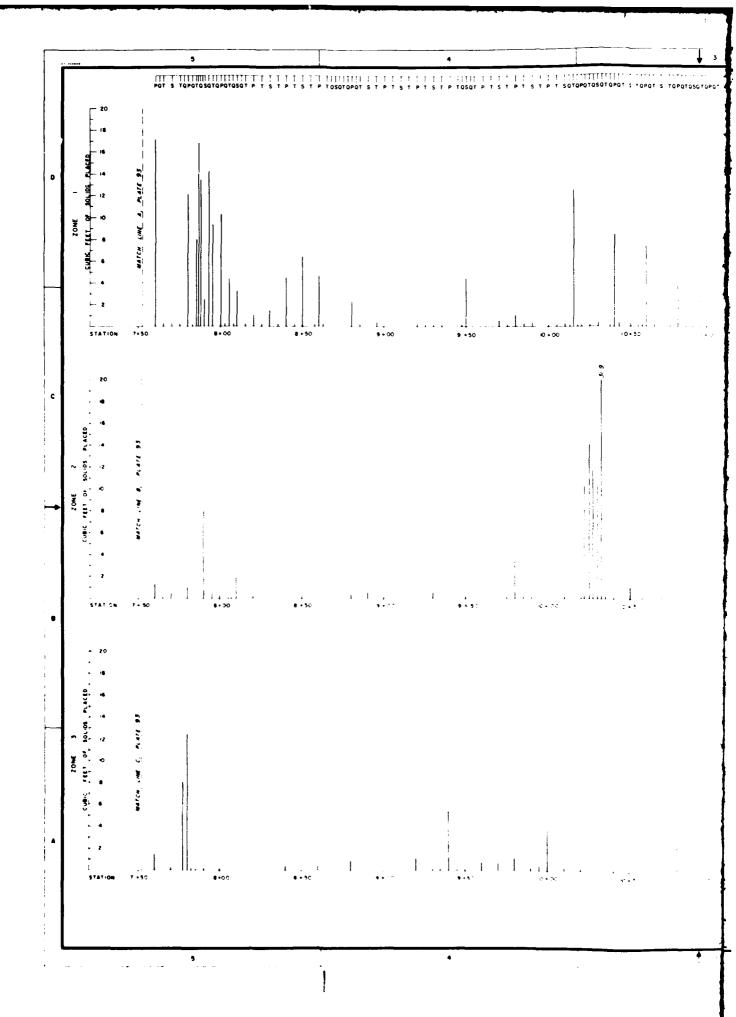


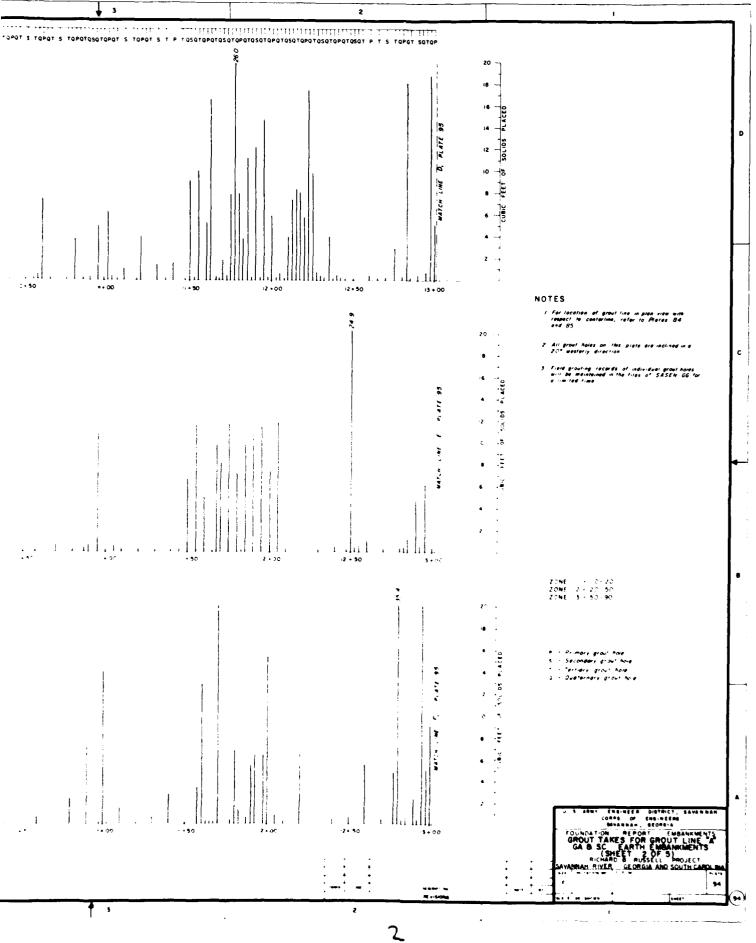


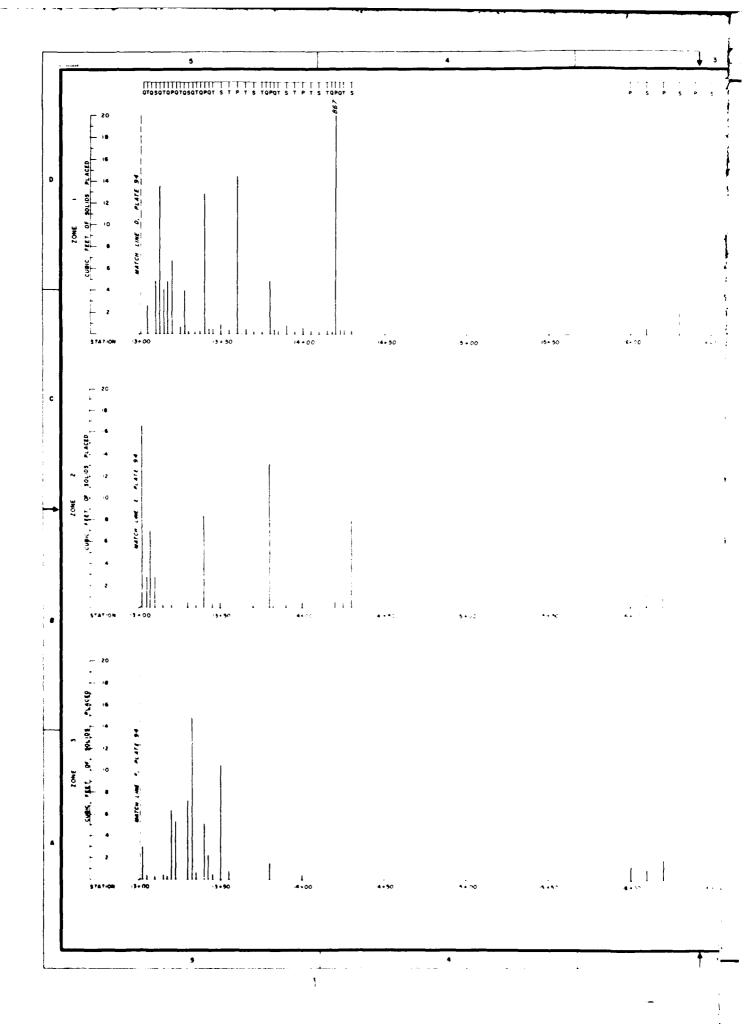




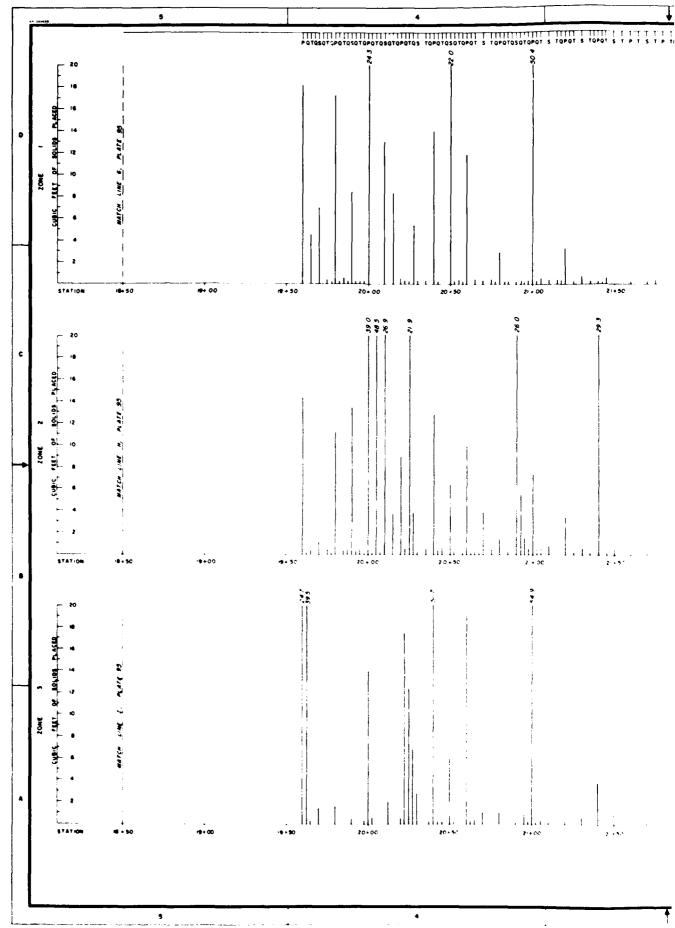


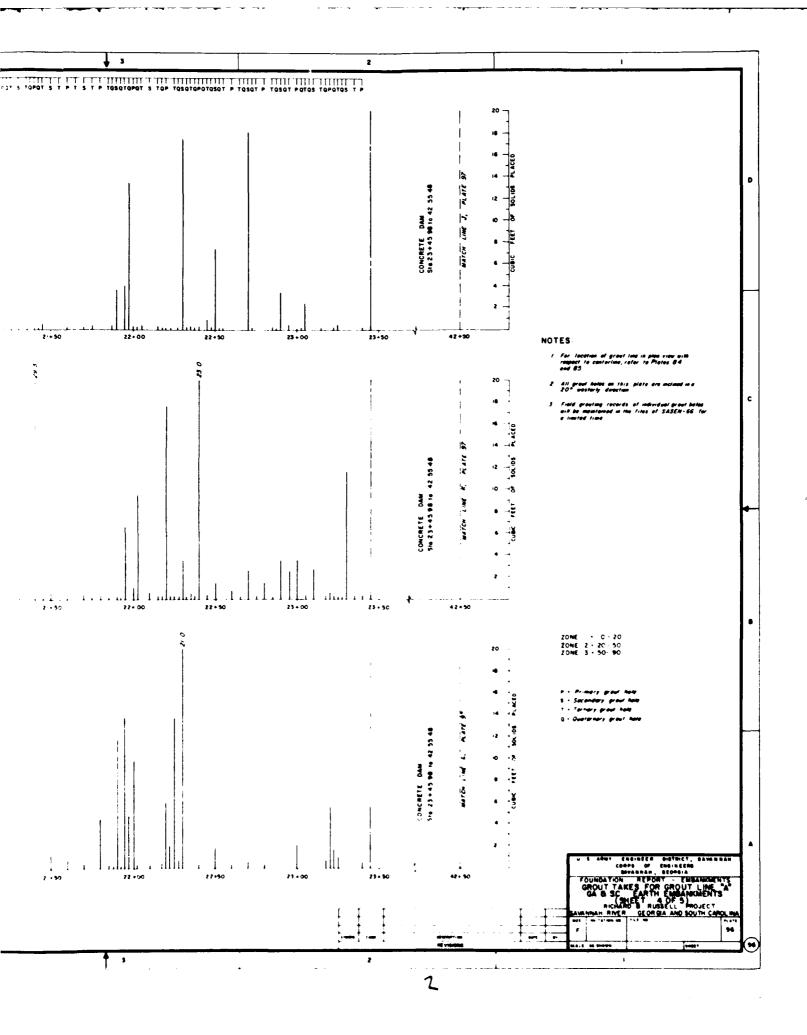


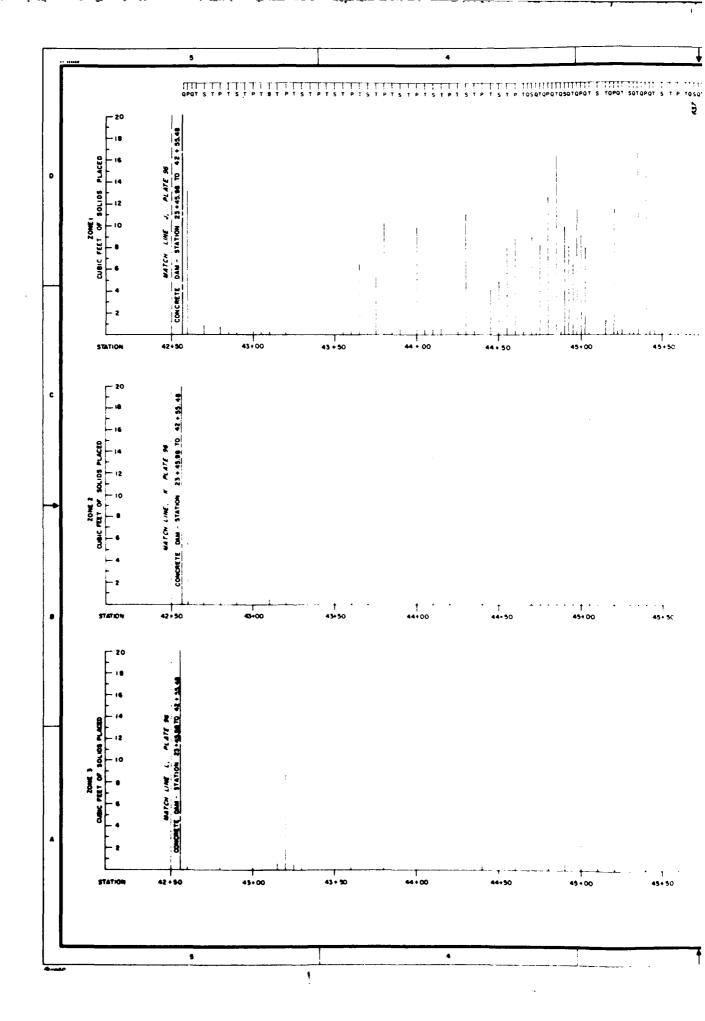


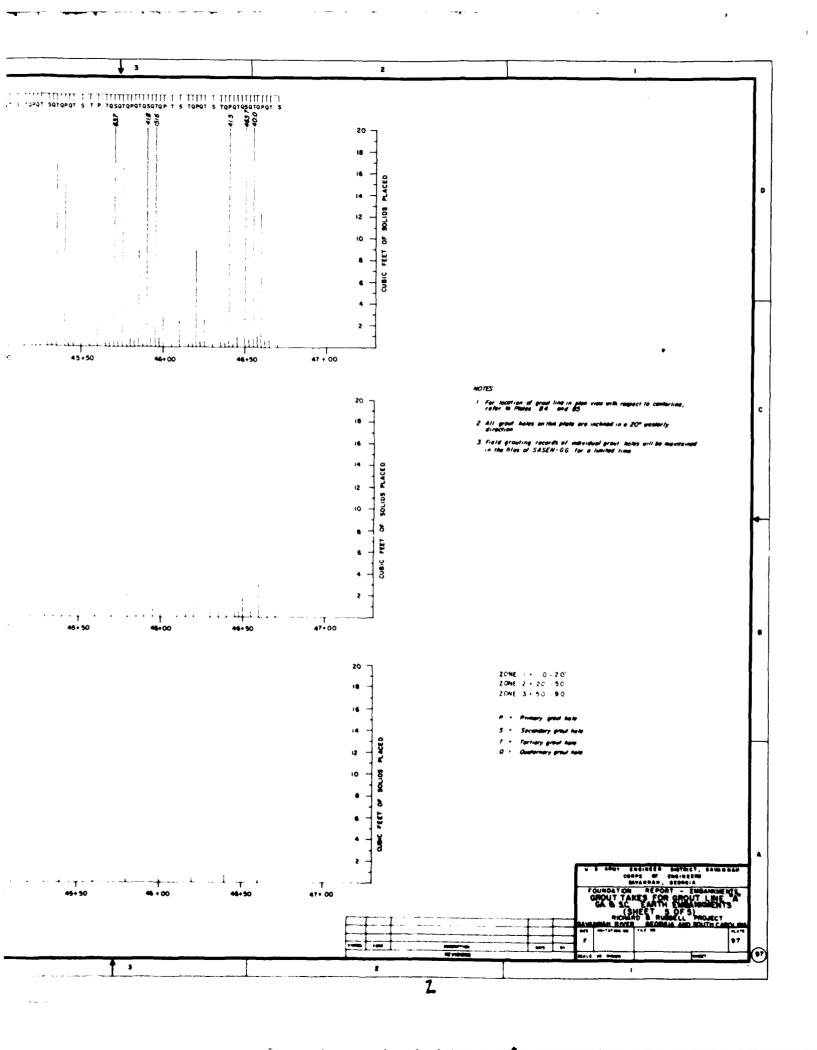


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				G PLATE 96	20 7 1 16 17 16 17 16 17 16 17 17 17 17 17 17 17 17 17 17 17 17 17
				WATCH CHE	6
5.50	•• 05	7+50	: 8 +≎♂	18 • 50	2 ·
					NOTES For location of grout time in a on view with respect to centerline, teller to Plates 84 and 85 2 41 grout hores on this prate are inclined in a 201 westerly direction Feld grout ng records of individual grout holes a 1 de martigined in the files or SASEN-GG for a mind time.
				9 22 2 2	6 - 20 - 20 - 20 -
9-1 7	** >c	*. 90	0.20	• • • • • •	20NE + 0 70 20NE 2 - 21 N 20NE N + 40 90
					20 a - Primary gract hole 5 - Secondary grout hole 5 - Tertary paut hole 3 - Judiernery grout hole 16 - G
				on 1, 9, 9, 1, 1, 1, 1, 1, 1, 1, 1, 1, 1, 1, 1, 1,	* *** * ** * ** * *** * *** * *** * ** *
				* 7.78	• • • • • • • • • • • • • • • • • • • •
4:50		.*****	18 - 00	8 + 50	TANKER DISTRICT, BAVARHAR COPS OF THE RELEASE FOUNDATION REPORT - EMBANGUENTS GROUT TAKES FOR GROUT LINE "A" GA B SC LEATH FURNINGENTS RICHARDS BUSSEL BROUEC' SAVARNAM PIVER GEORGIA AND SOUTH CARD, MA MAI 1 1 1-000
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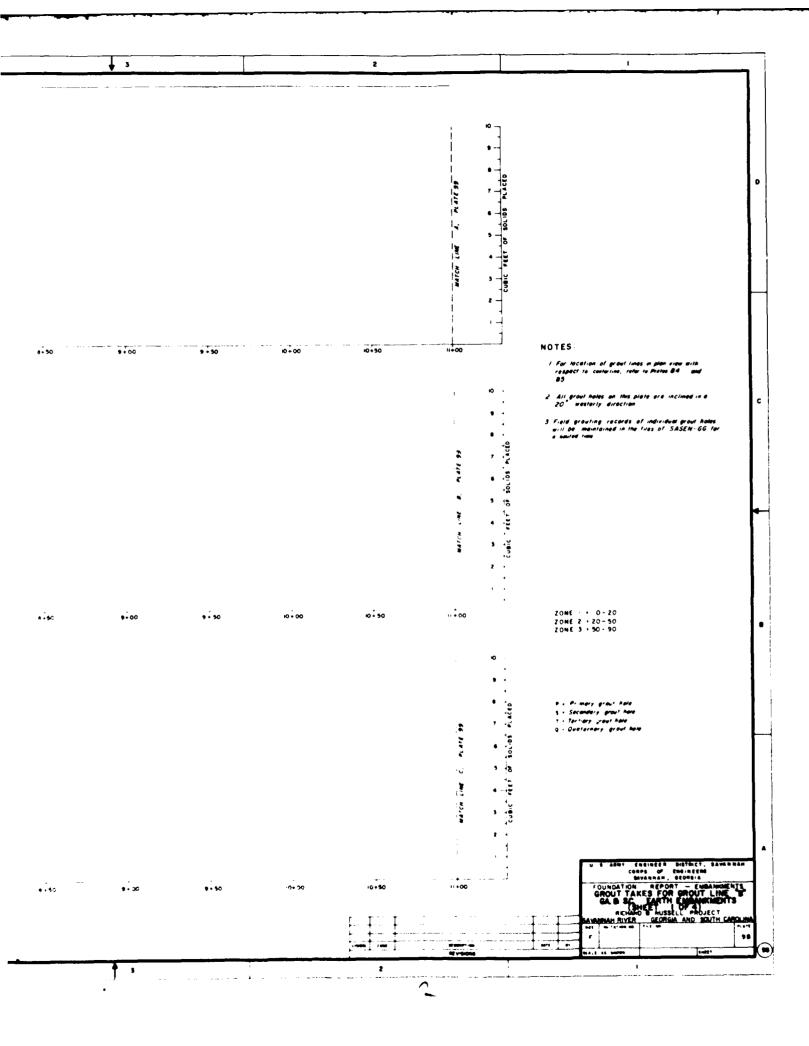


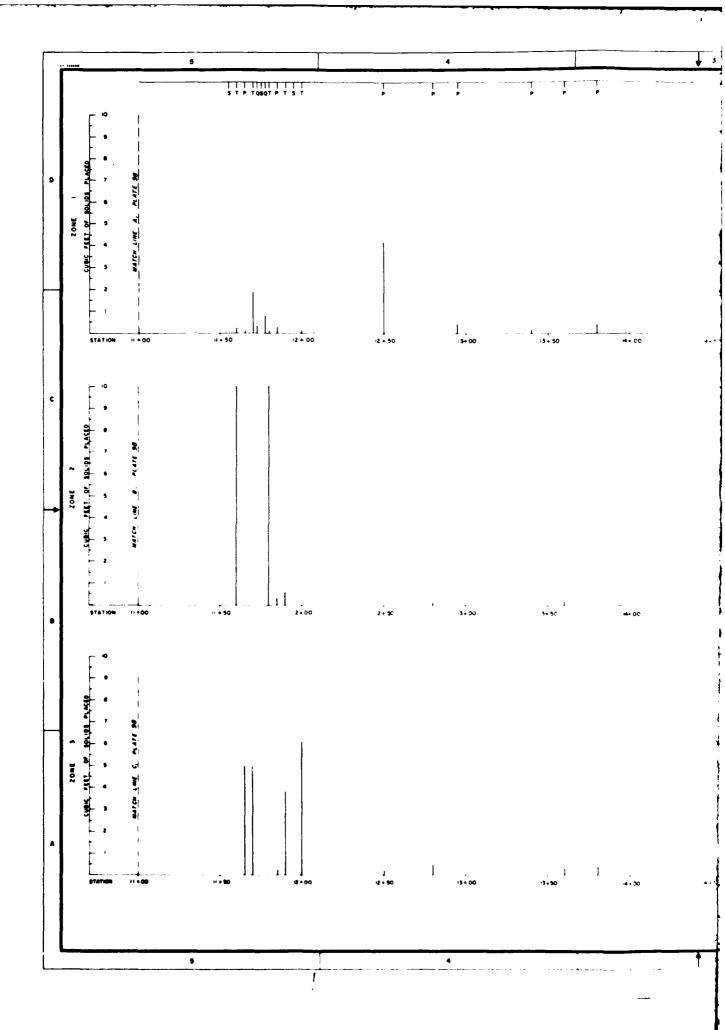


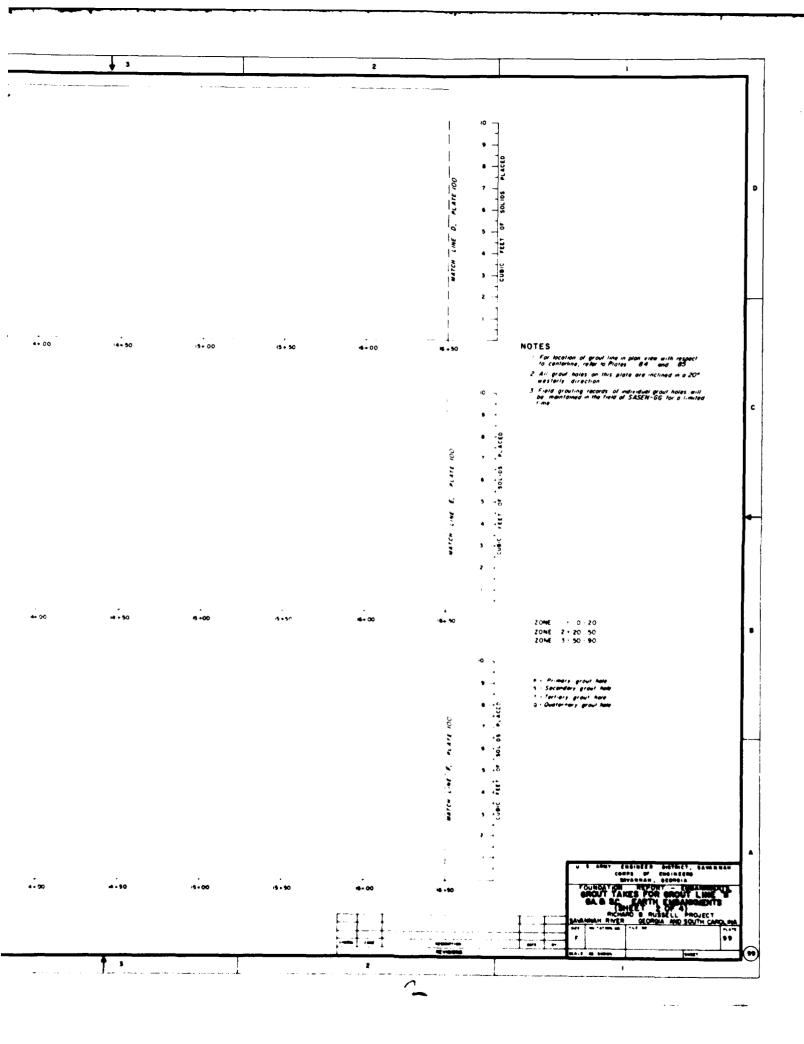


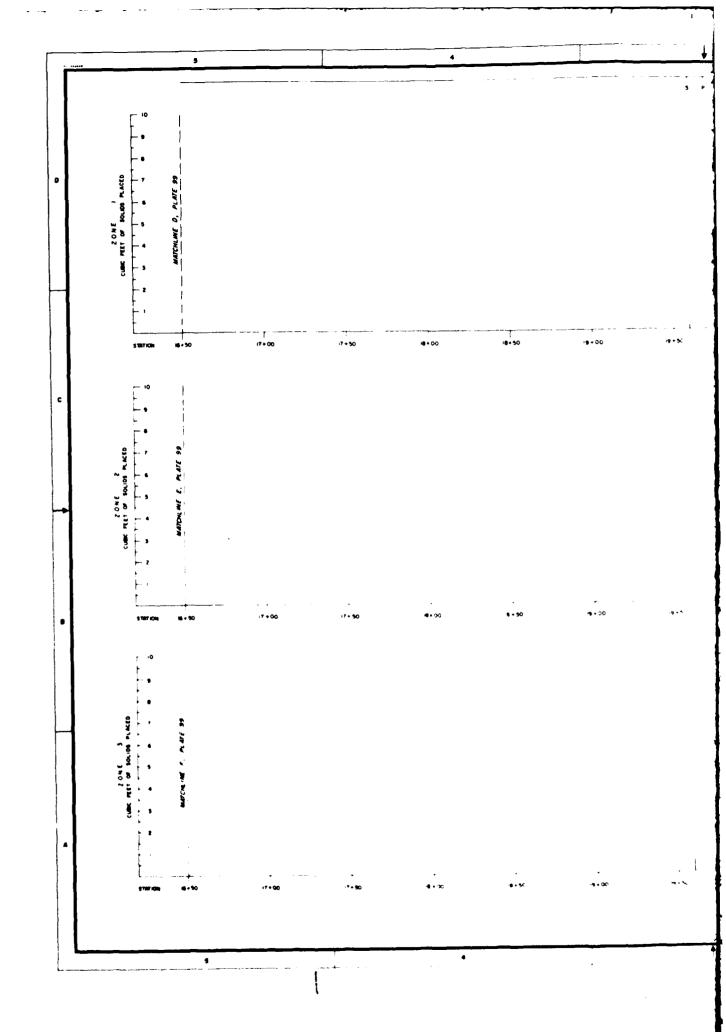


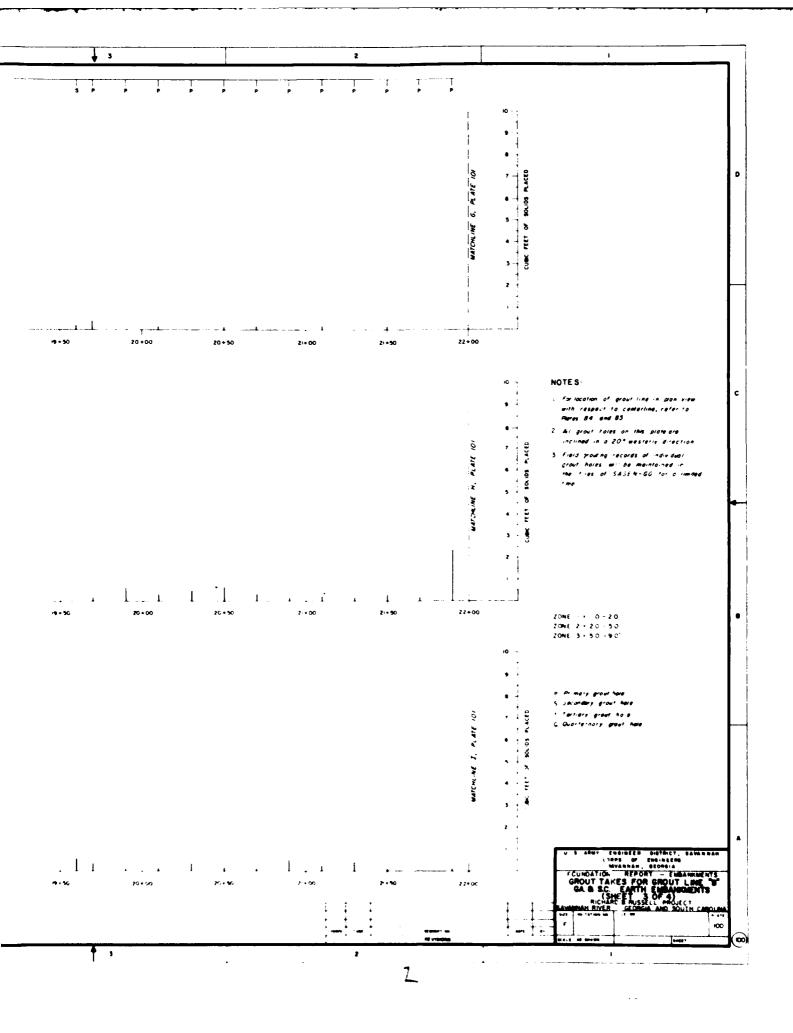
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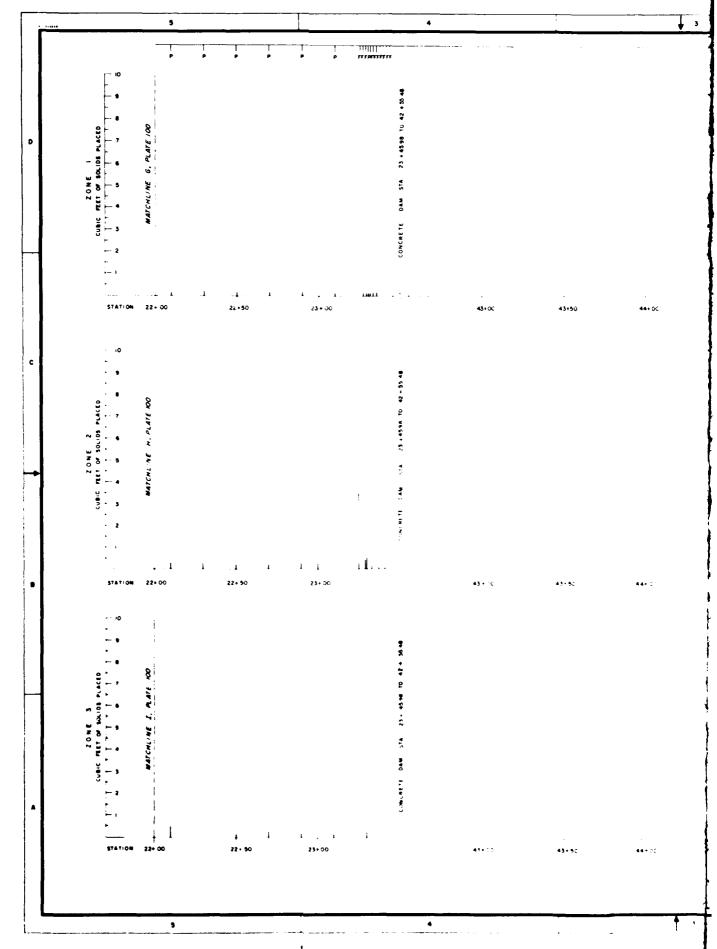


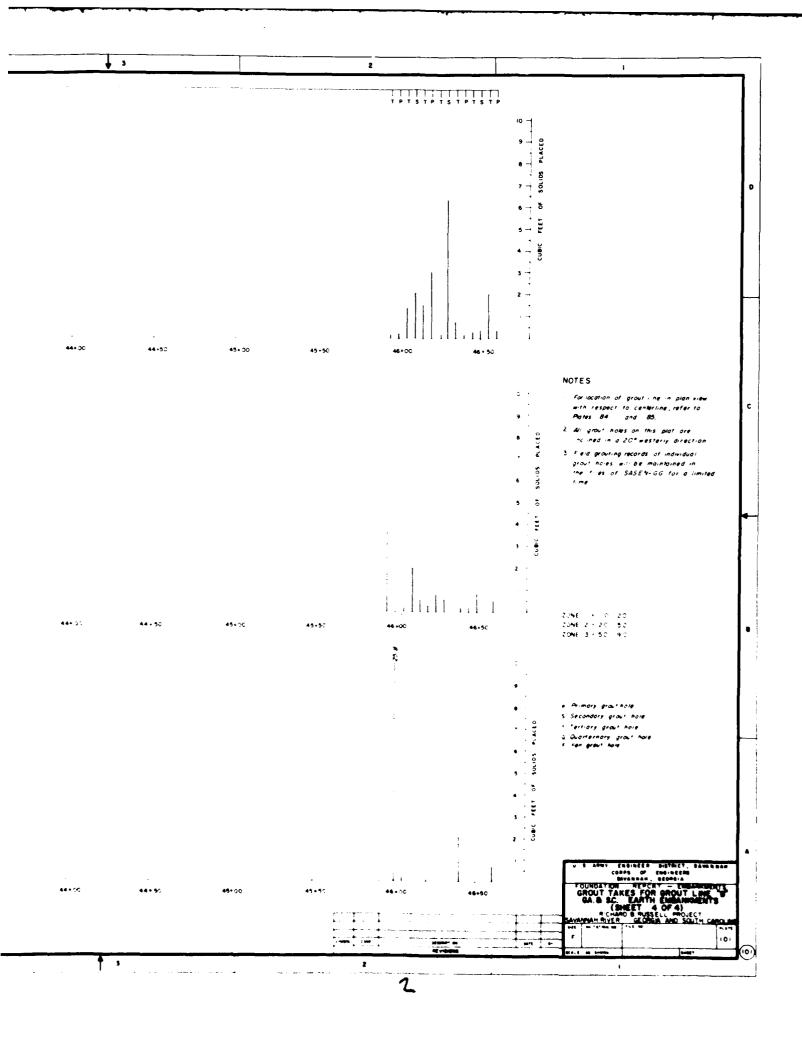


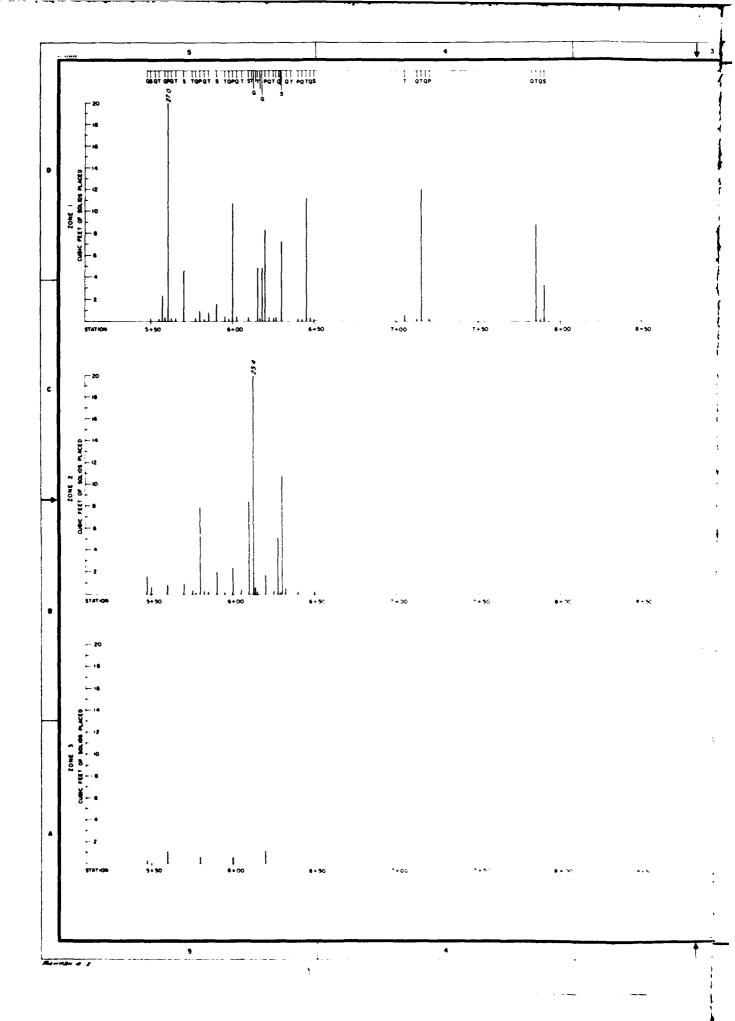




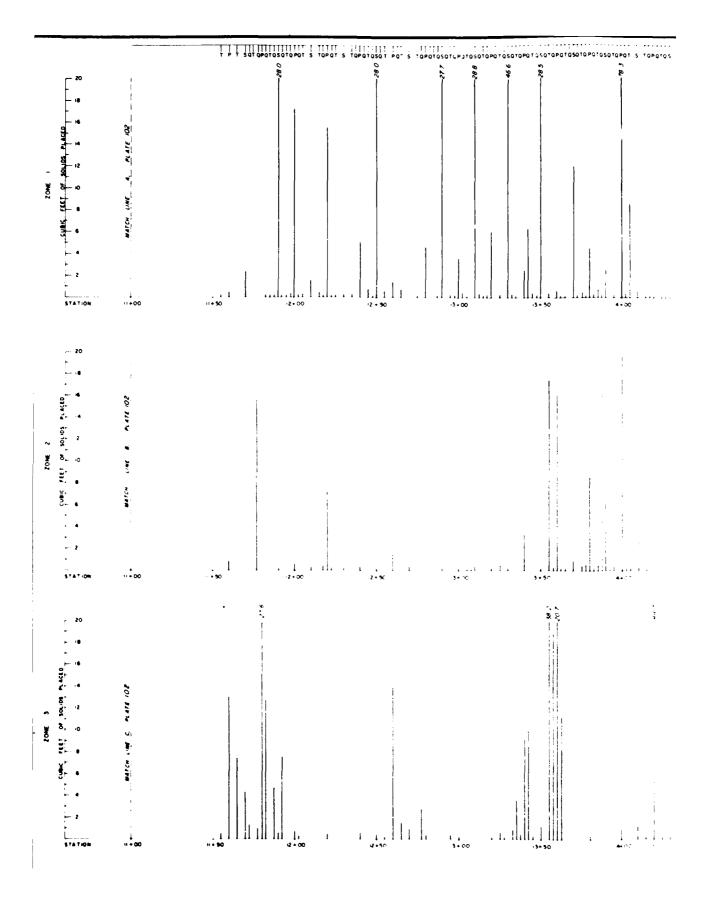


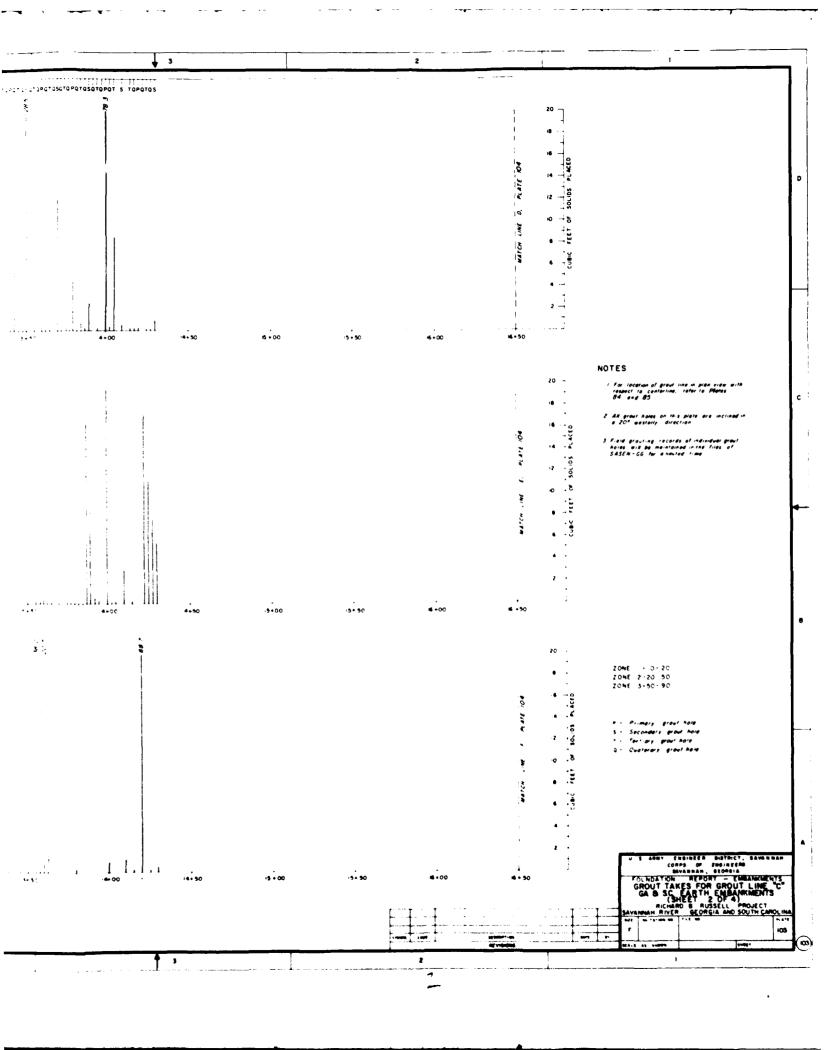


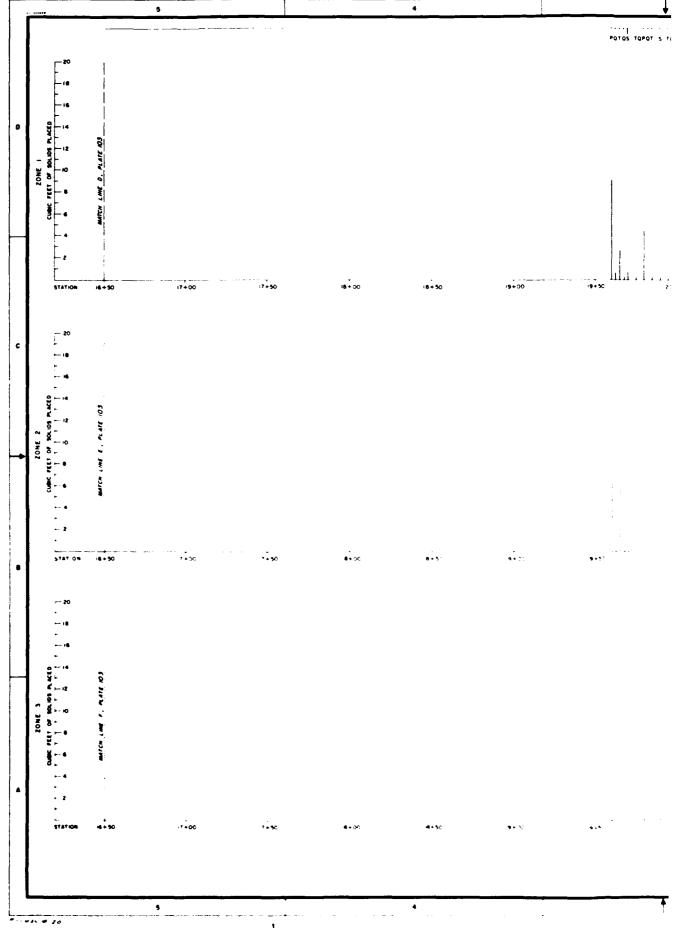


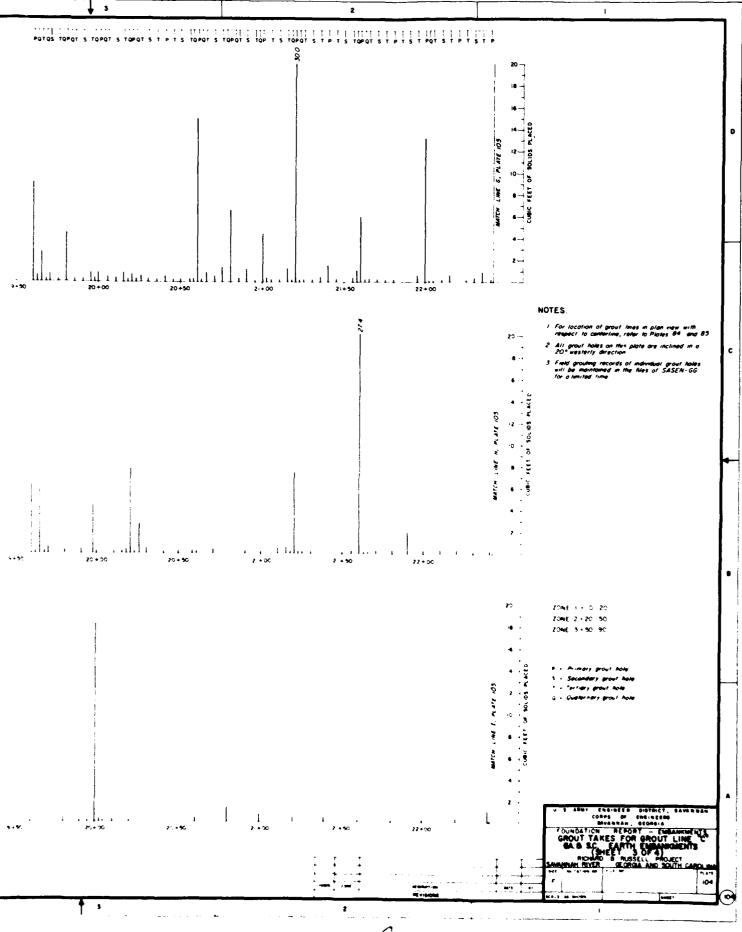


3 2 20 --MATCH LINE A. PLATE 103 9+00 9+50 10+00 11+00 8 - 50 NOTES I For location of grout lines in plan view with respect to Centerline, refer to Photes 84 and 85 20 -2 All grout holes on this plate are inclined in a 20° westerly direction 3 Field growting records of individual growt holes will be individual in the files of SASEN-GG for a limited time. o ~ b MATCH LINE B. PLATE 103 z . • 00 5 • oc 0+50 ZONE - - - 0 ZO ZONE 2 - 20 50 ZONE 3 - 50 90 50 P. i. Pr. more, grout have 5. Secondary grout have 1 - Tertiary grout have 3 - Deatermary grout have COPS OF SHORE OF STREET, SAVARRAM
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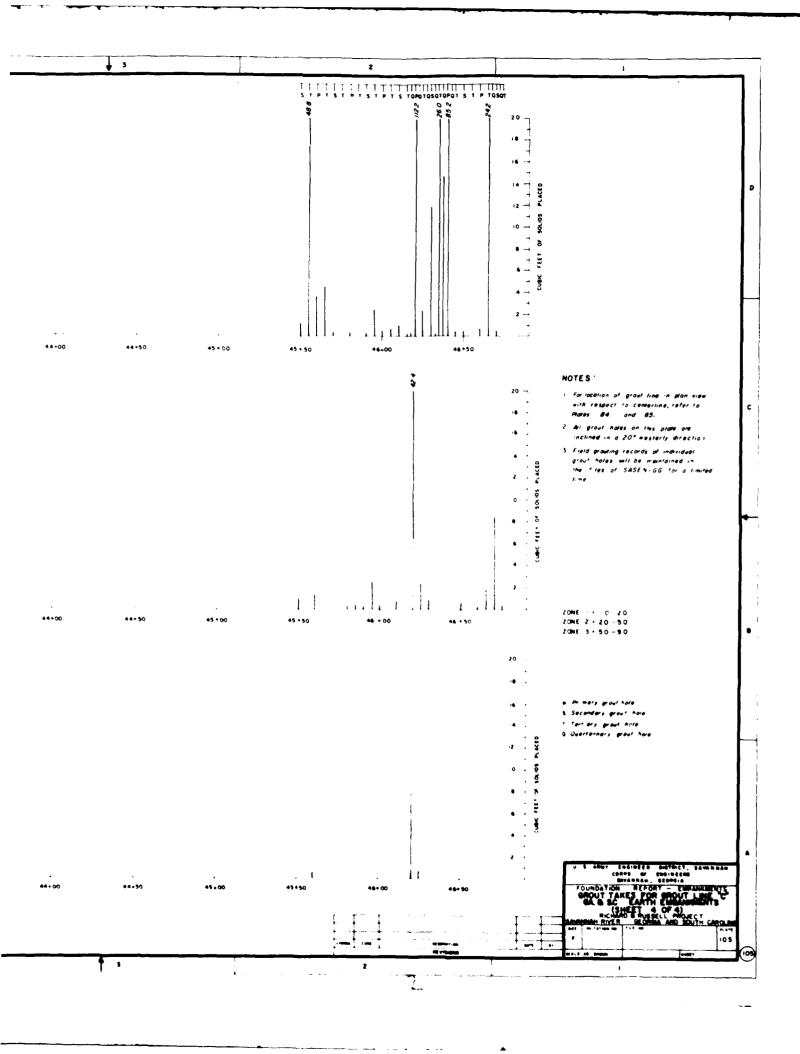


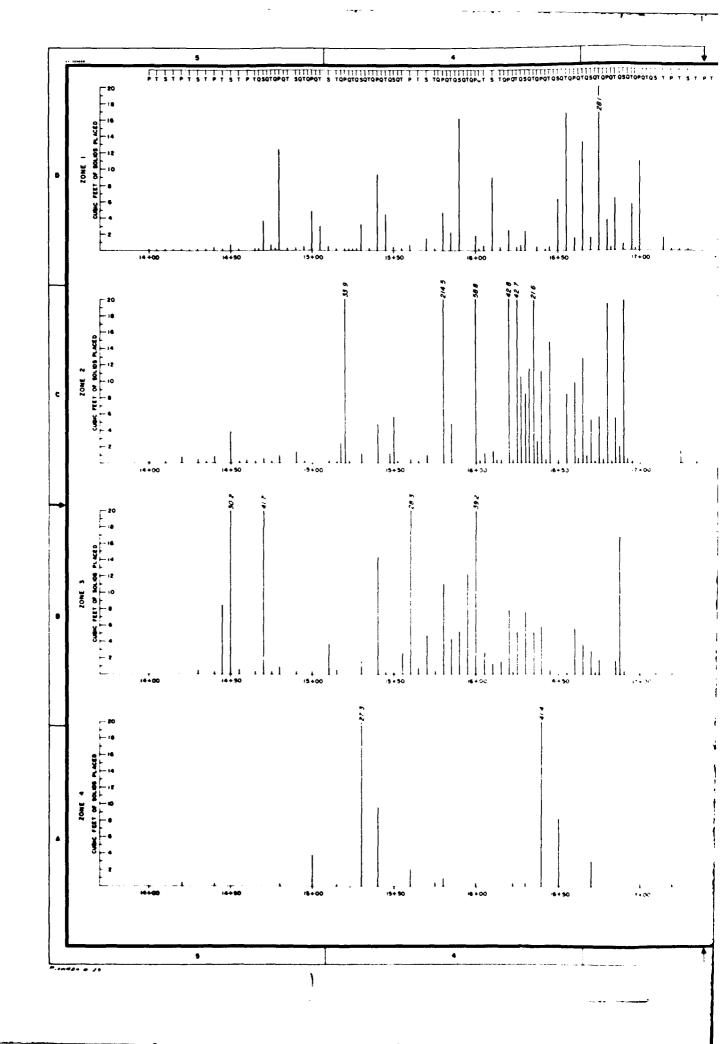






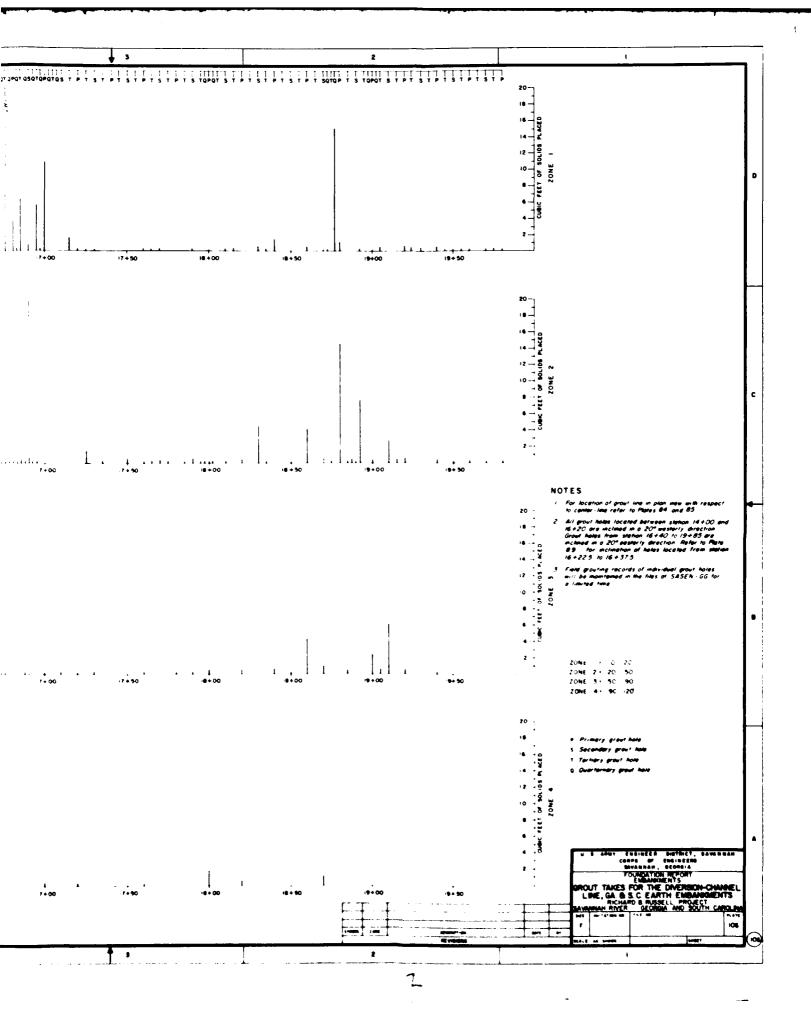
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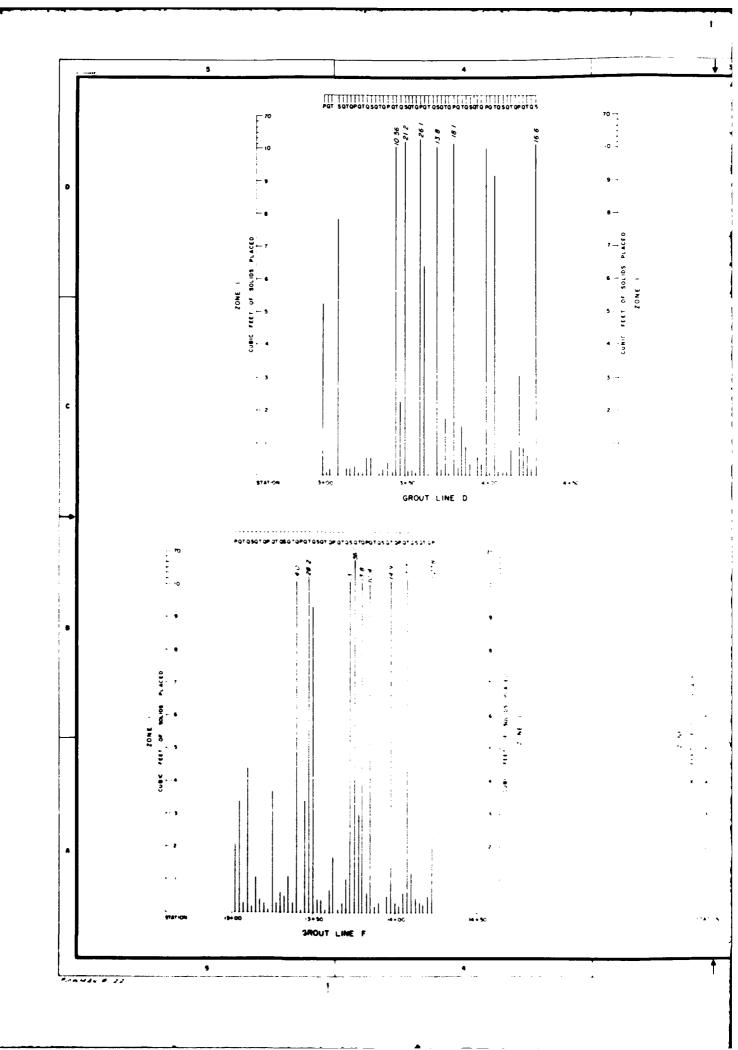


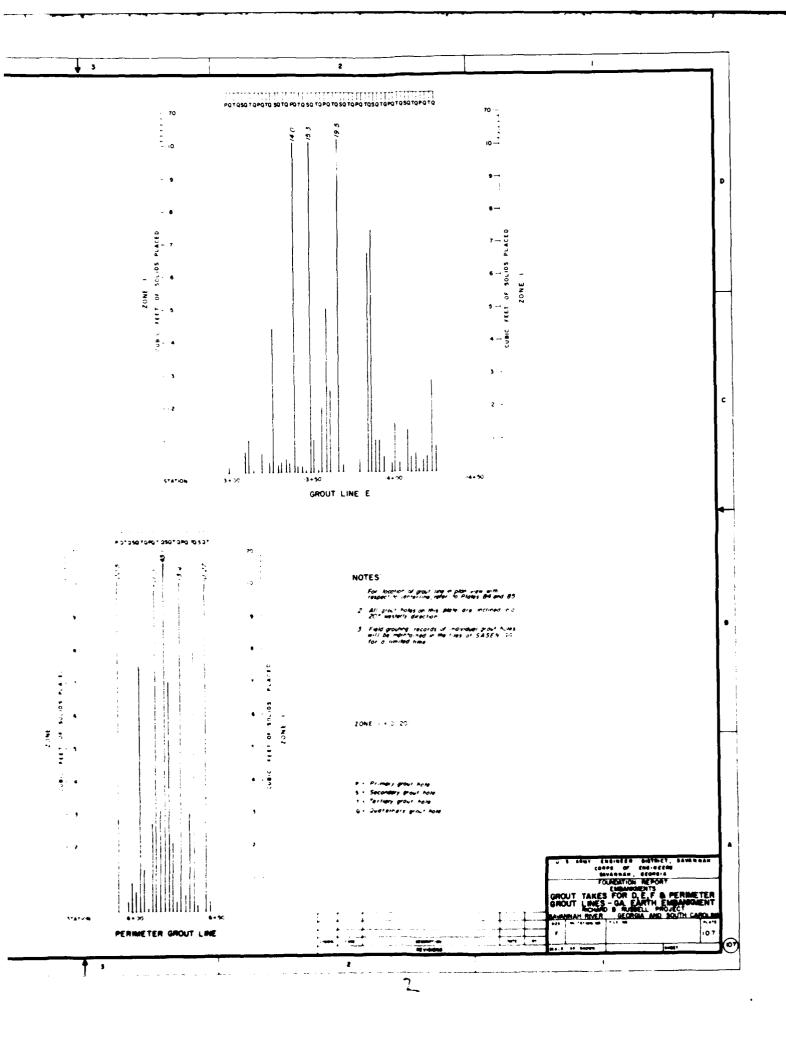


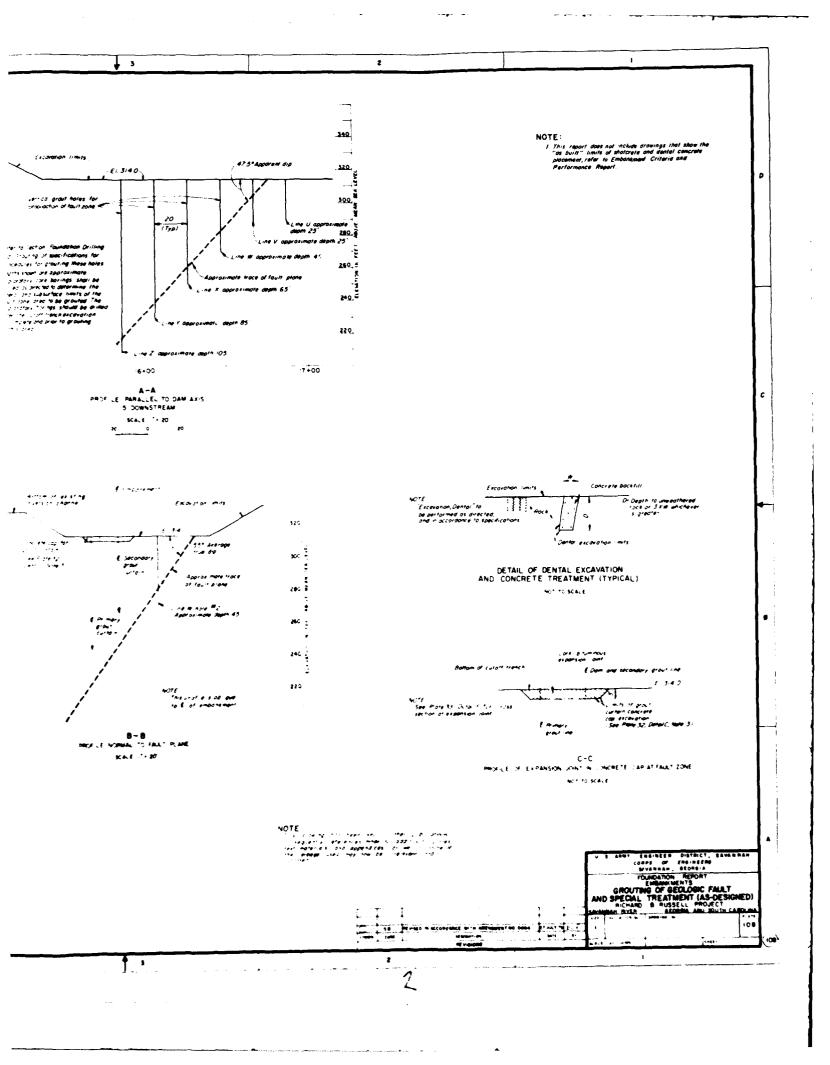
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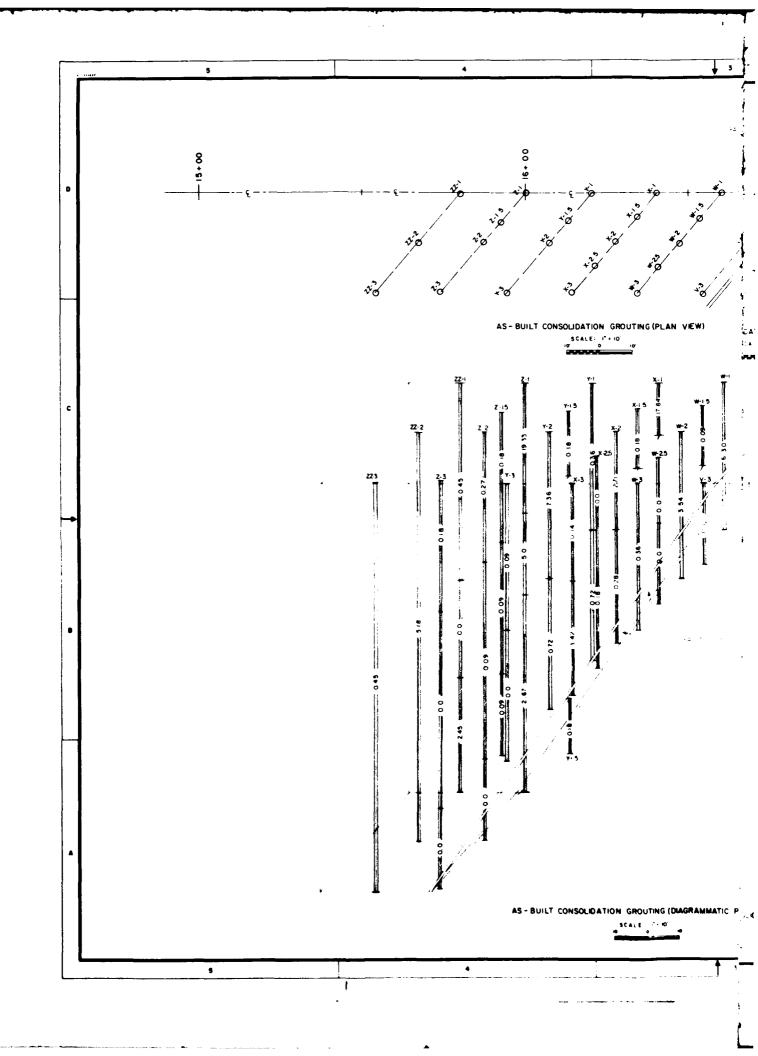
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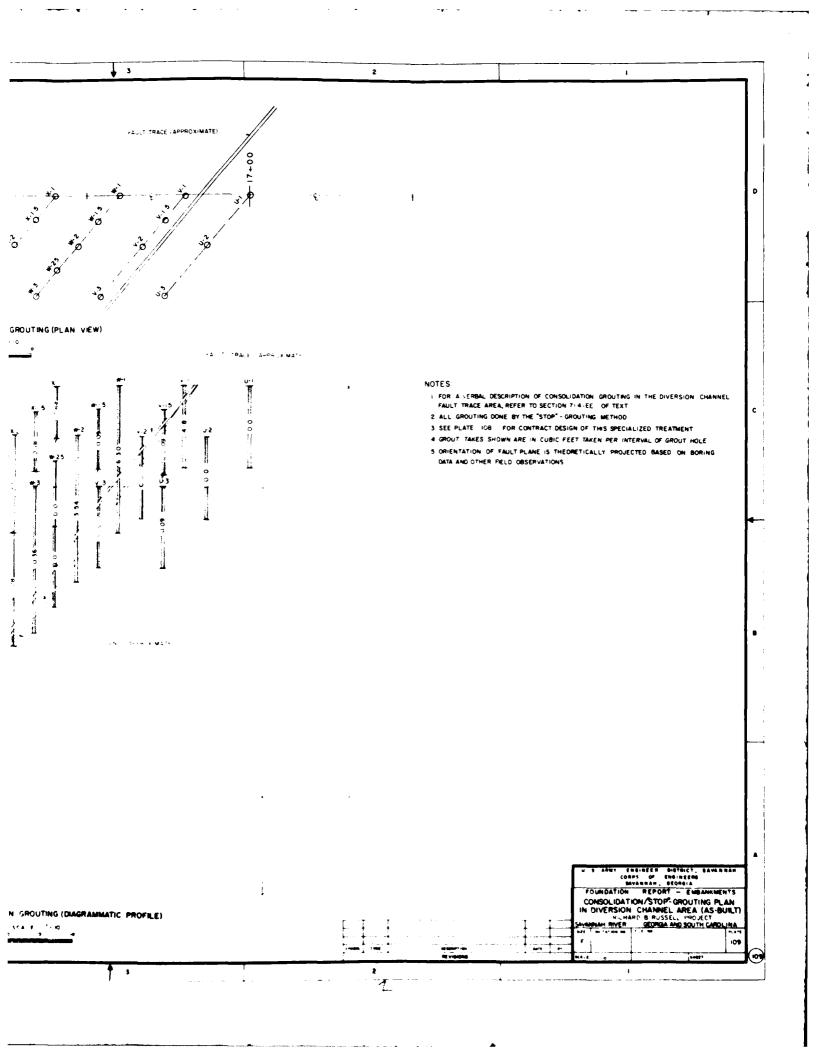




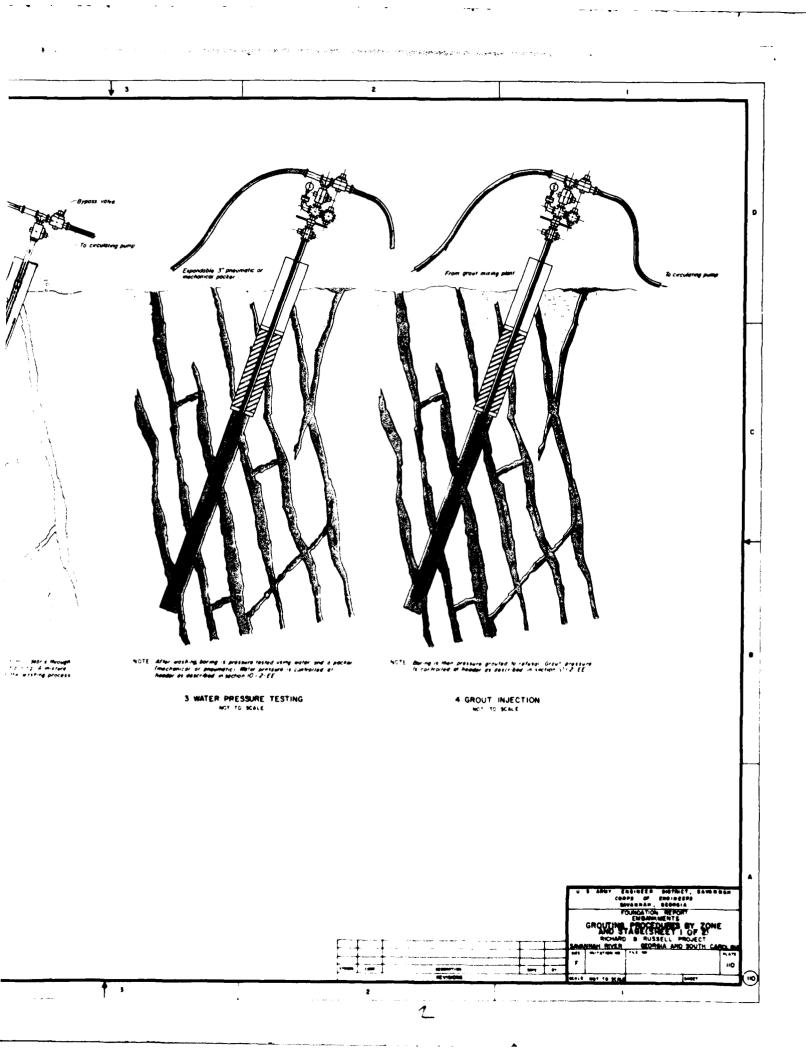




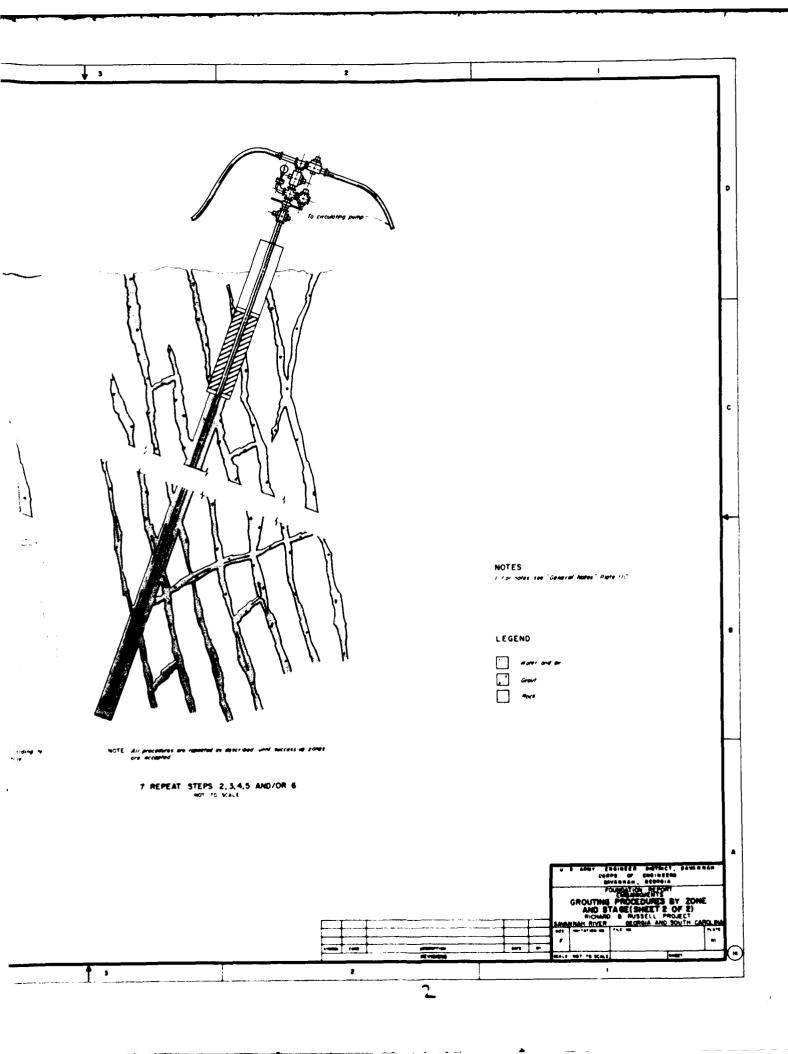


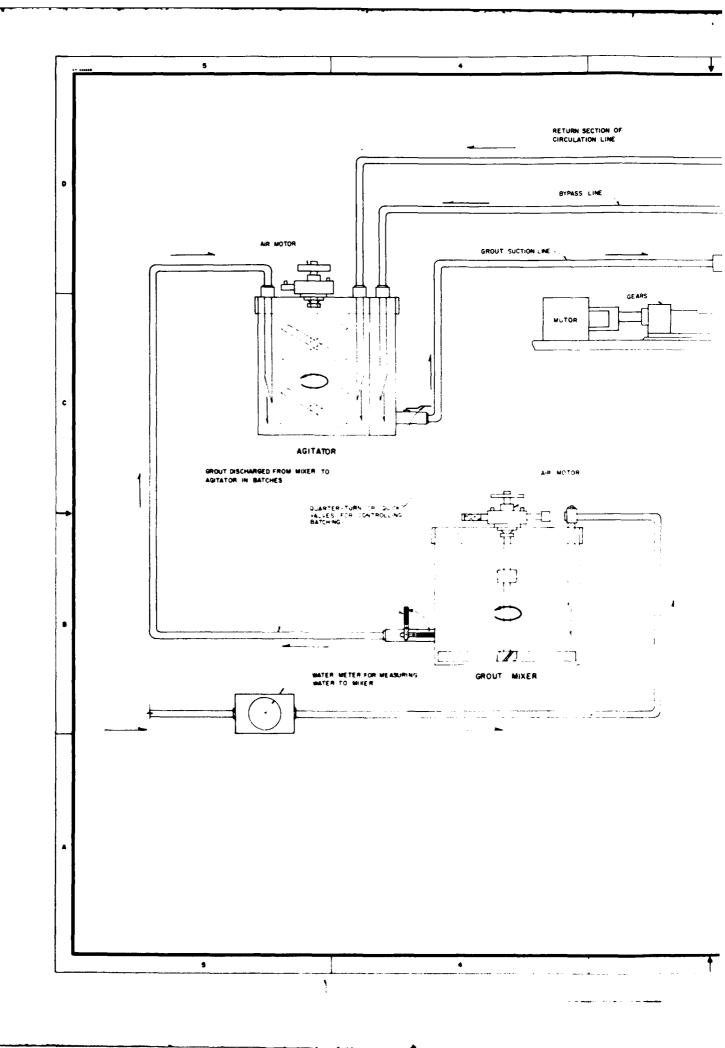


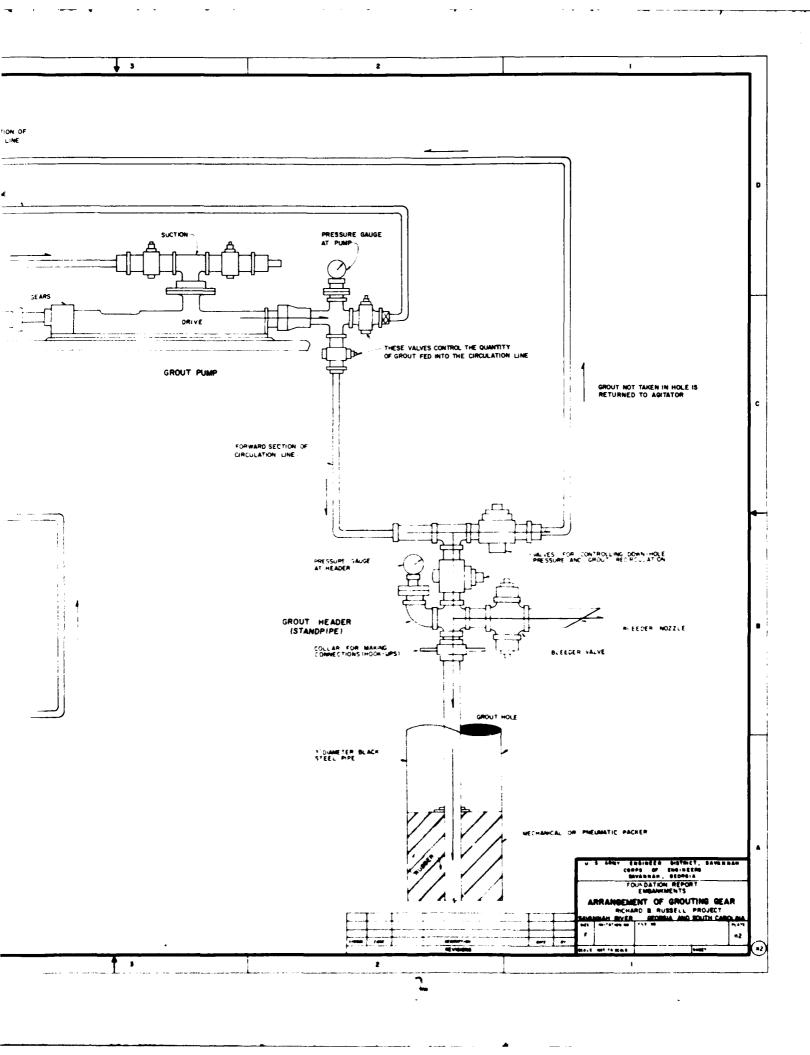
5 J-80" Approximate average dip of set I fractures See Piete 5 (is horse need a MOTE Boring is first distant to depend usually sone is using tolors and percession methods. A mistime of six and eater serves as a disting medium. NOTE: Boring is main worked of our rings and as it debris through an generated pupe owend to bottom of boring. A misters of complement on and water is used in the washing profess. I DRILLING 2 WASHING LEGEND GENERAL NOTES I Fracture size specing and extent the gross's exaggarited for purpose of Hhaffeton Set of fractures die Mose hosing the largest occurance throughout the project area. Refer to Plate (5)Communication between growt helps and surface leakage occurse randomly throughout the growting incosts, records of which are membered on help growing raports. B for improve purpose, these growings show decised peretration of agent and group into rock posits.

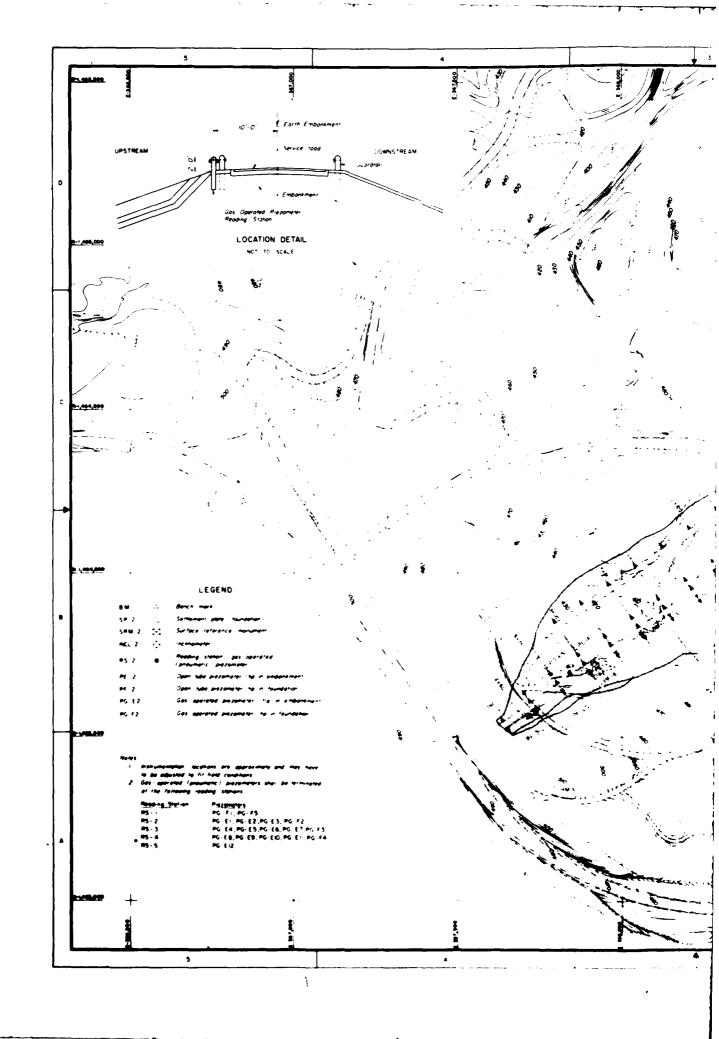


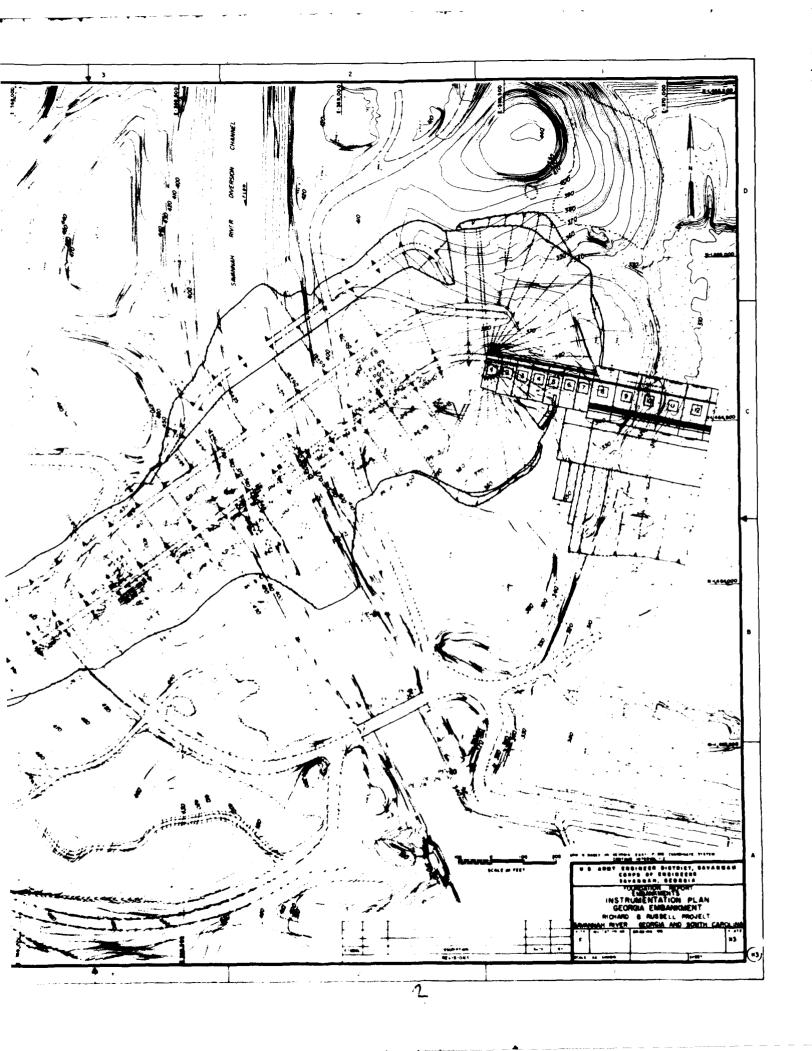
5 5 CURING 6 FLUSHING AND RE-DRILLING 4

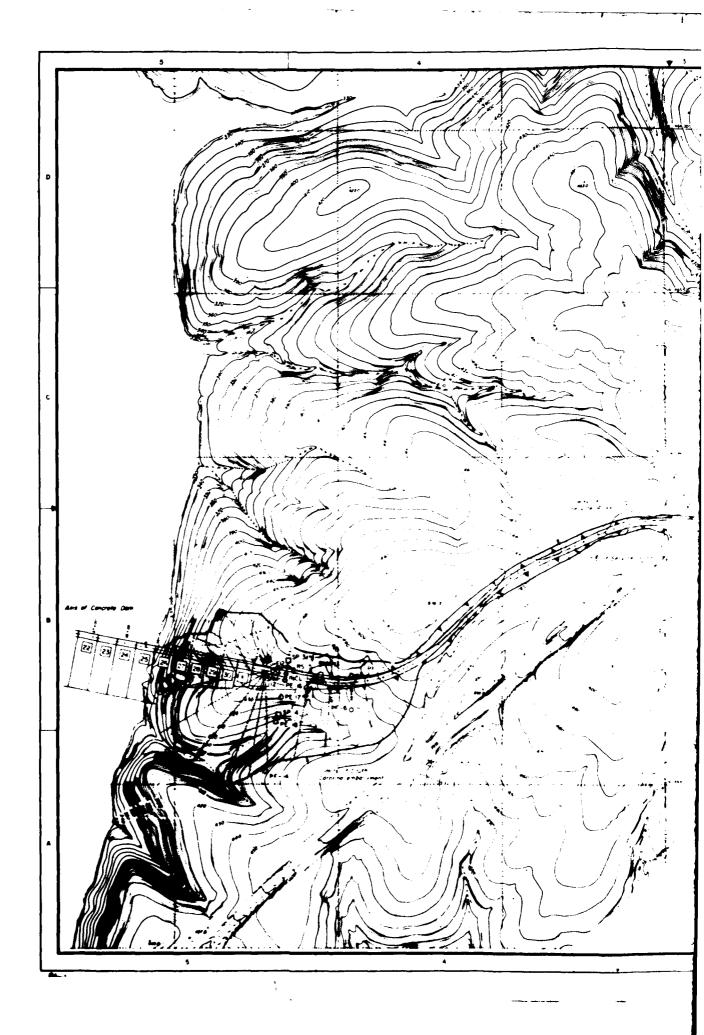








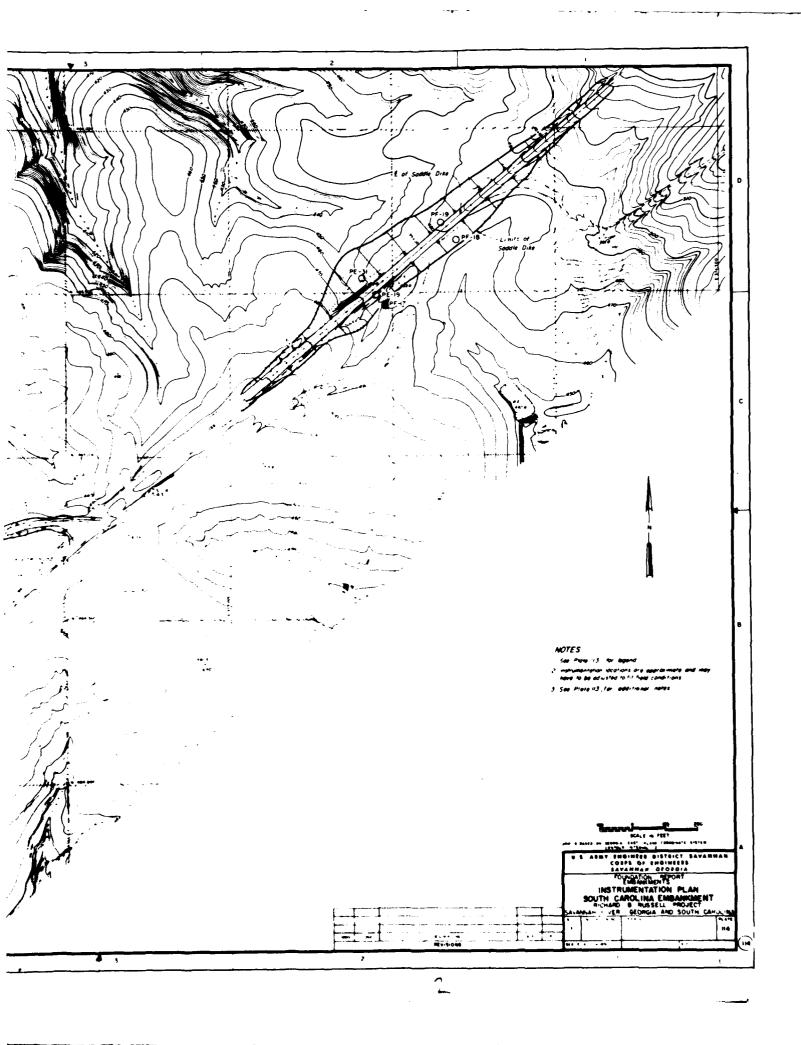


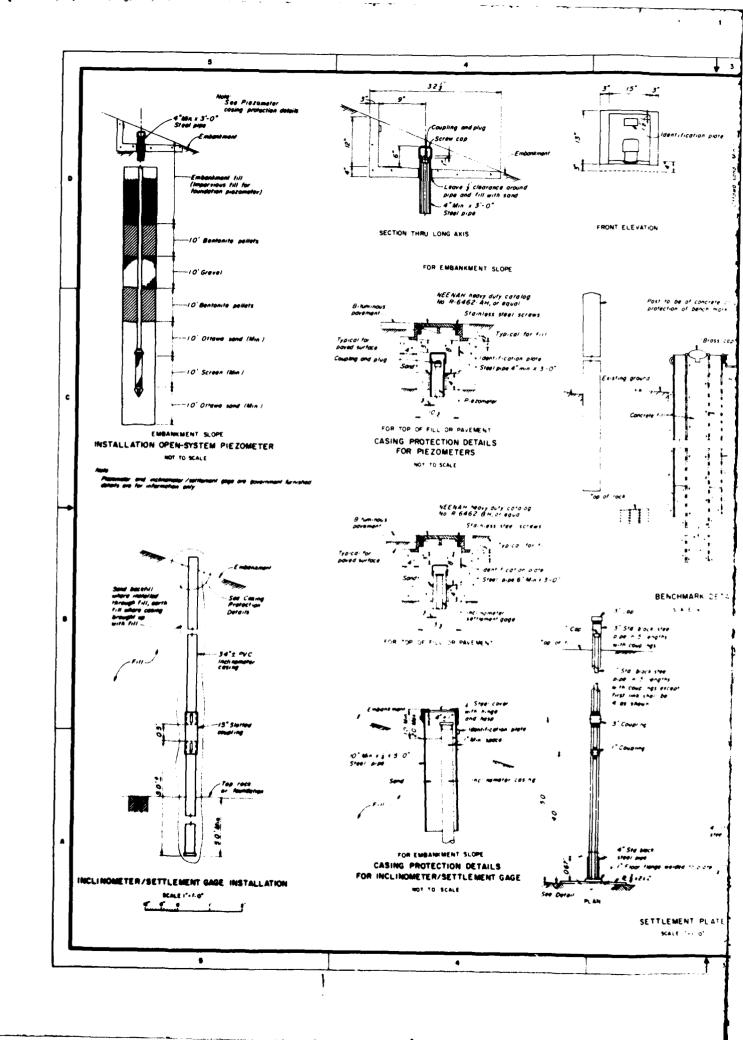


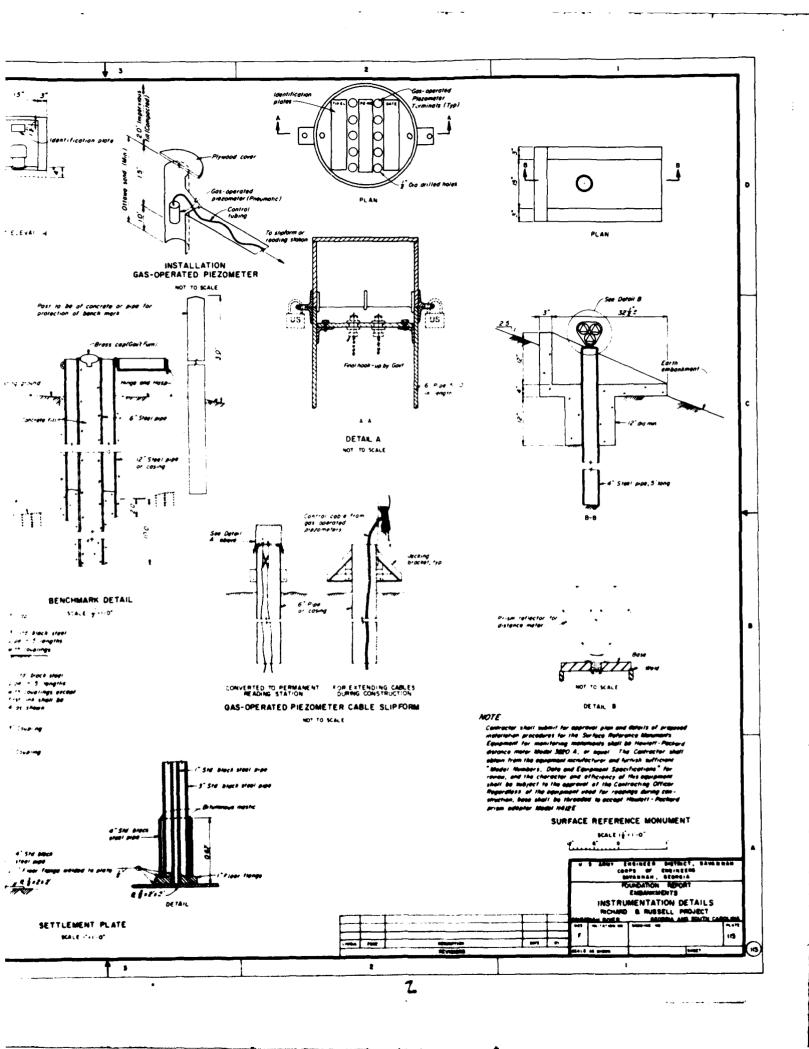
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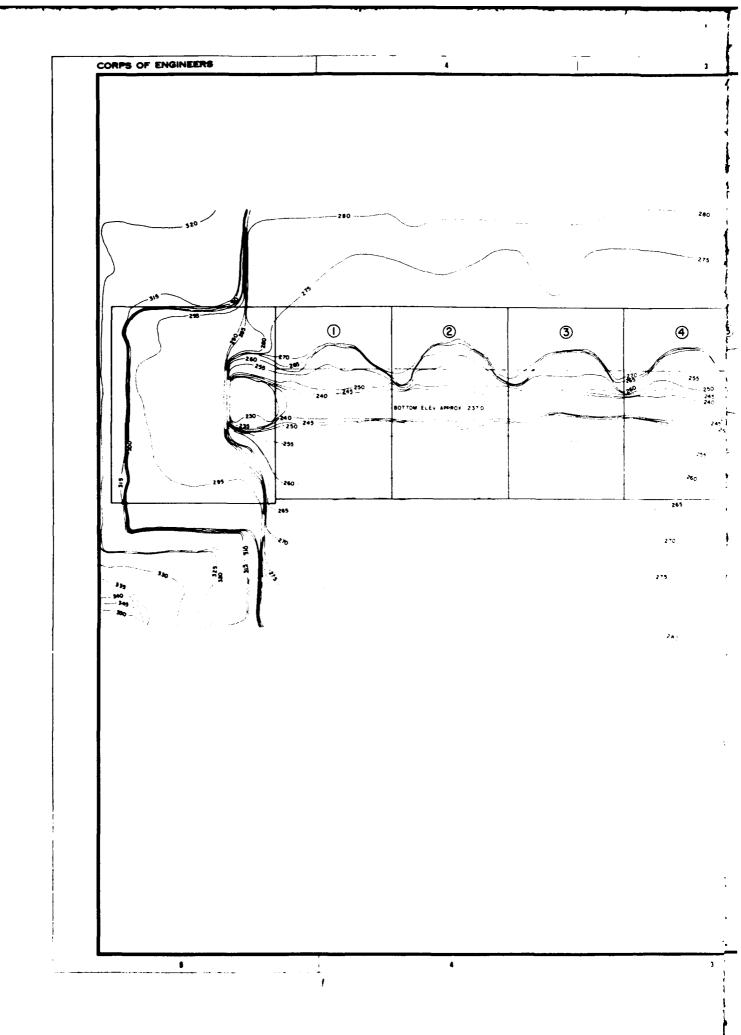
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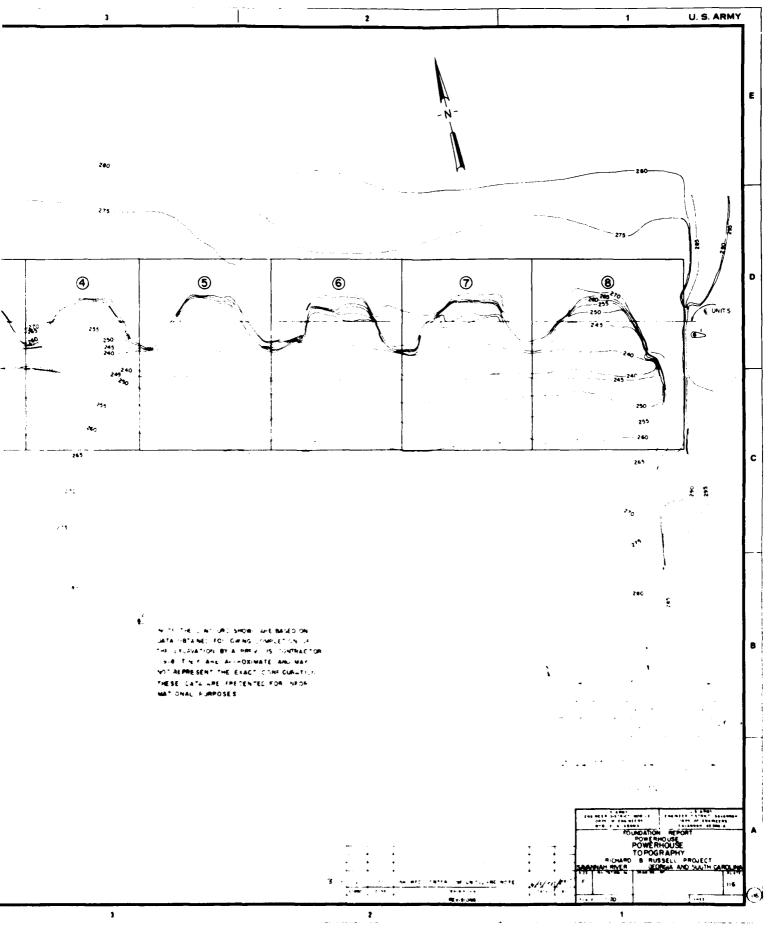
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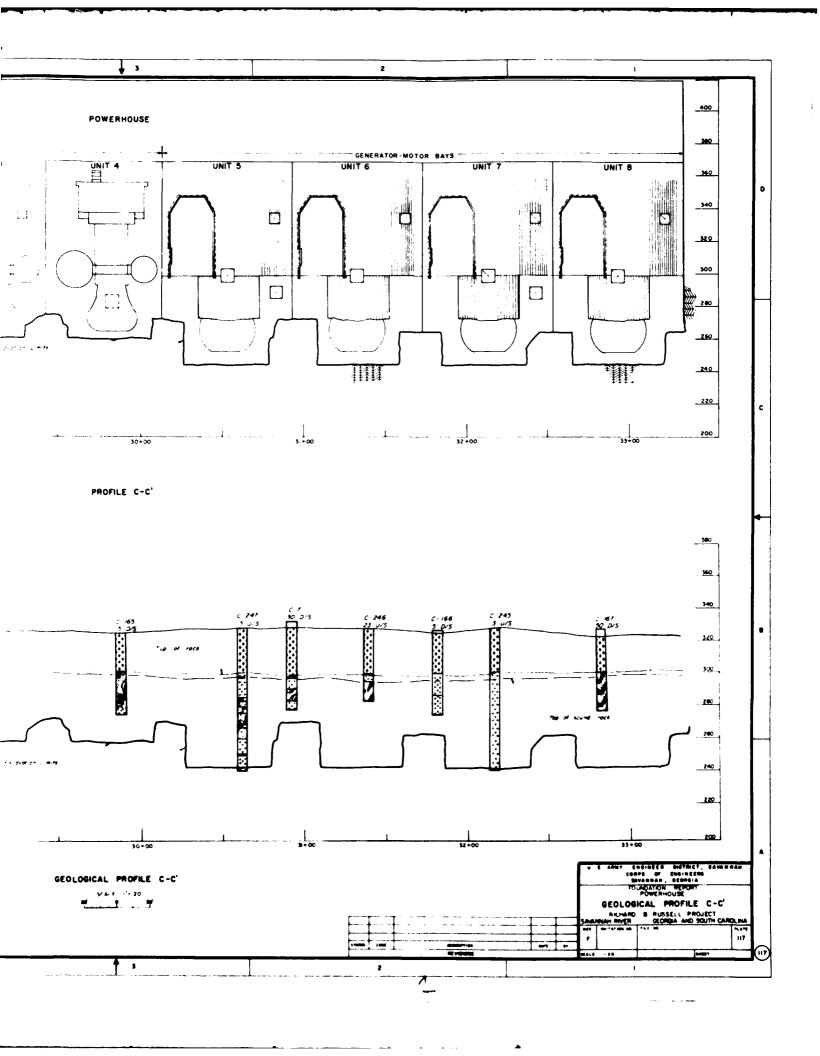




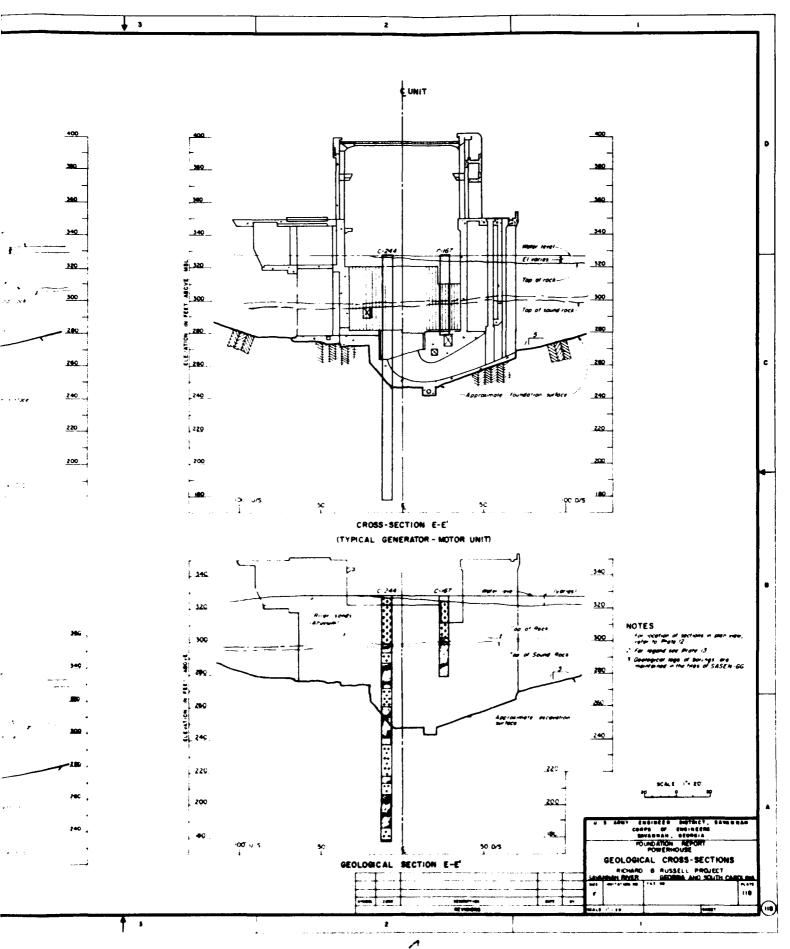


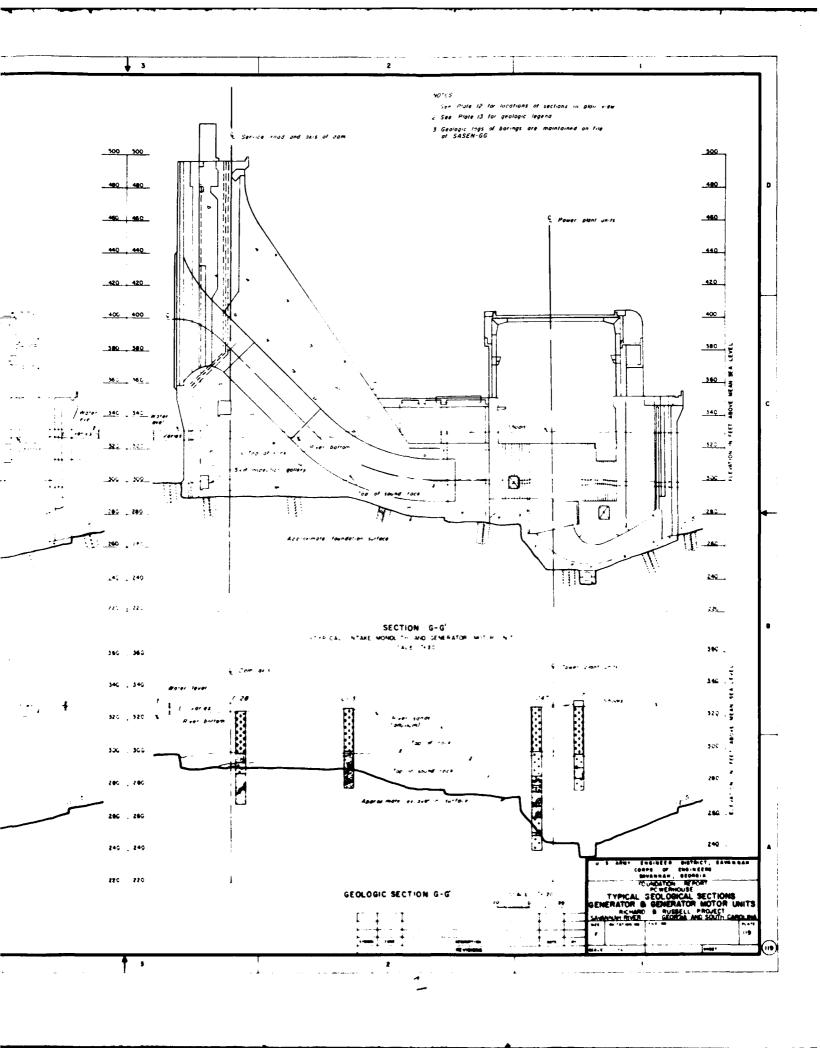


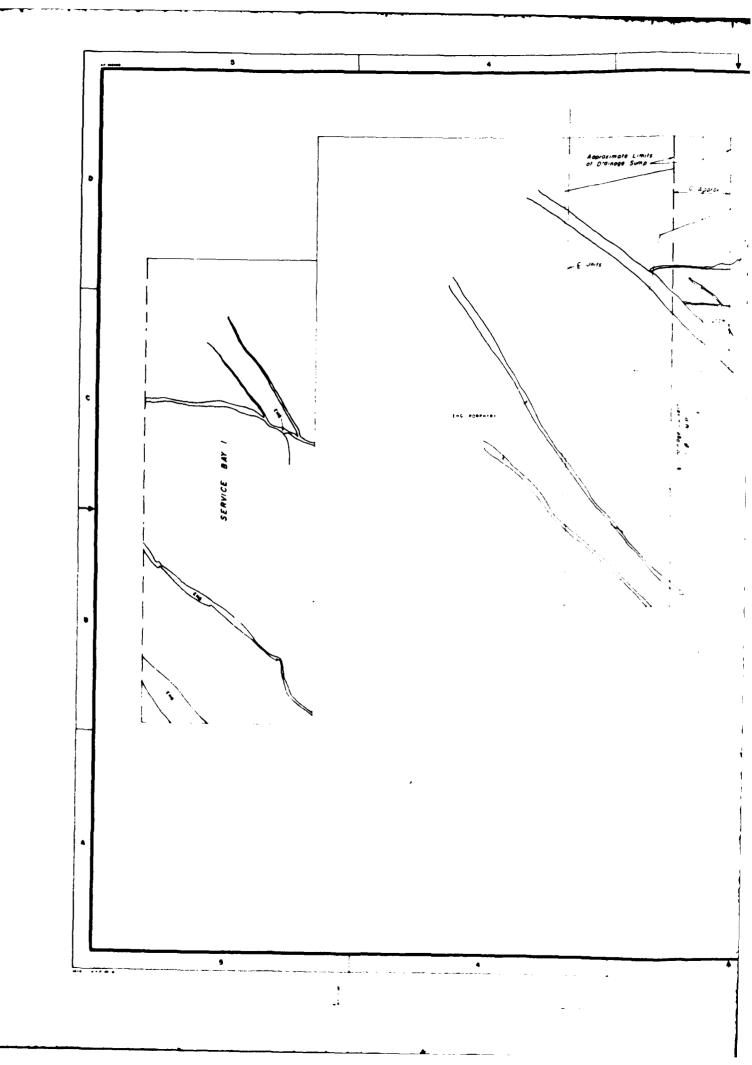


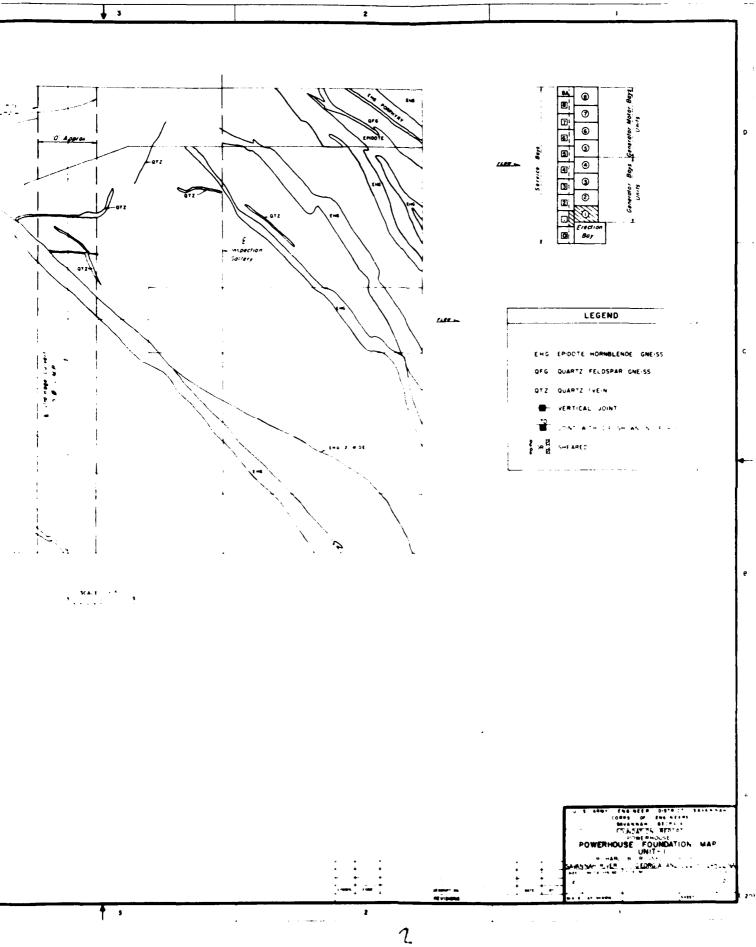


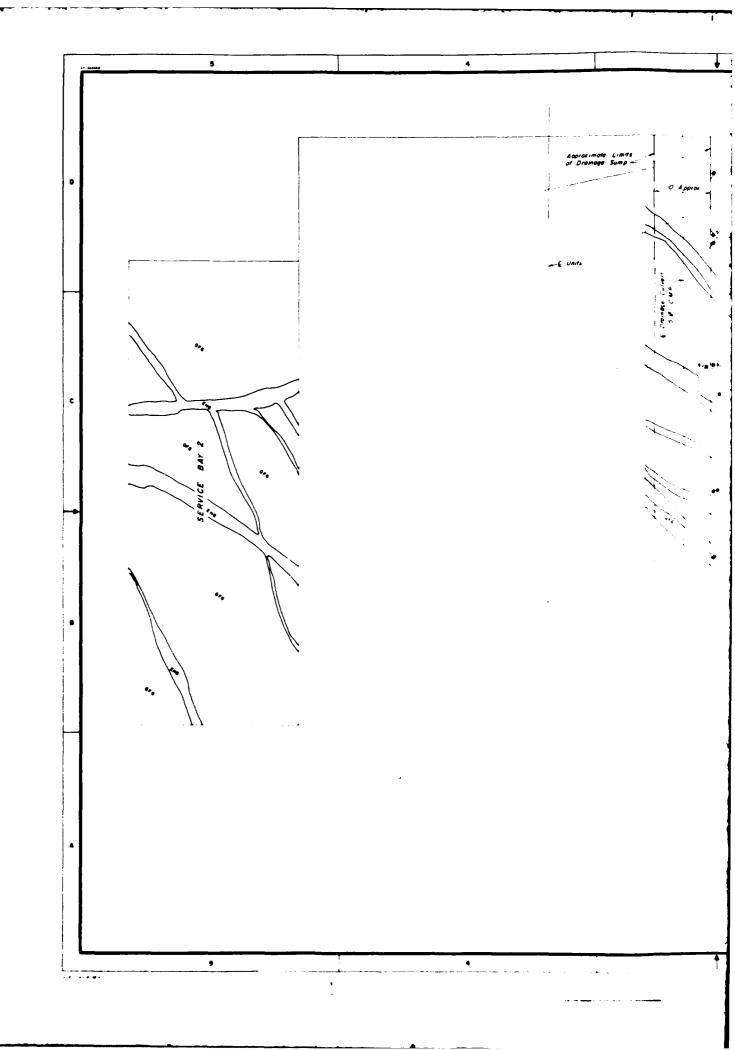
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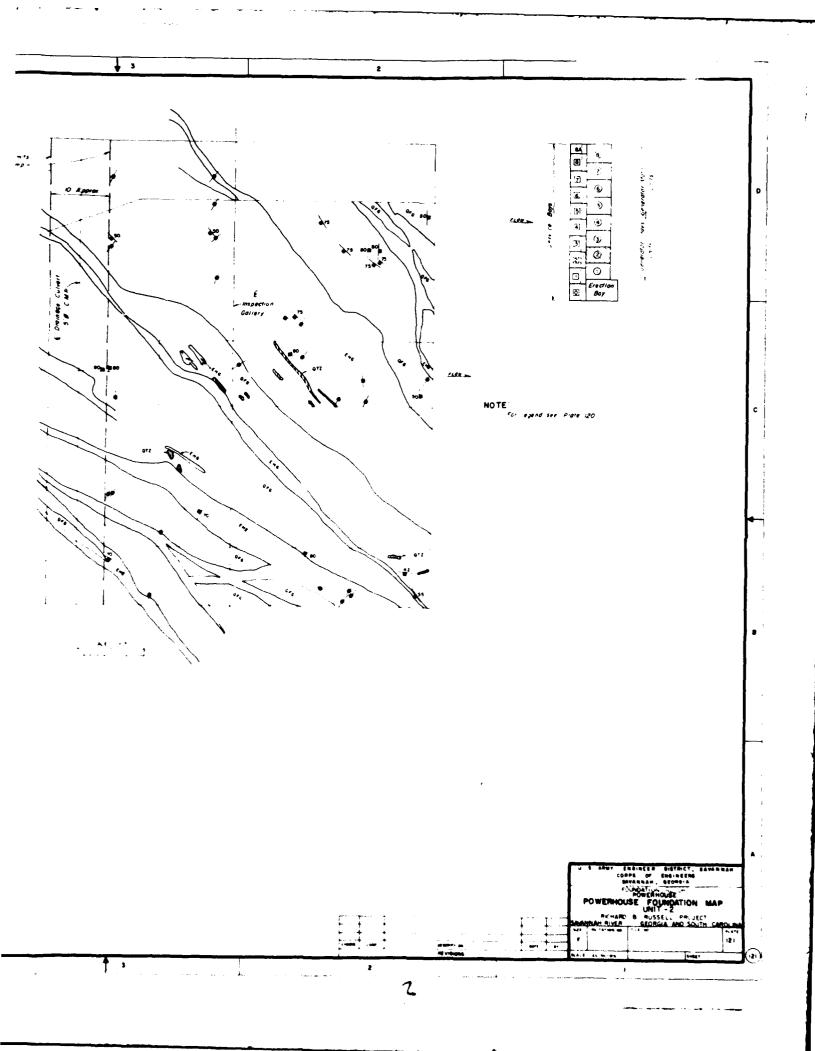


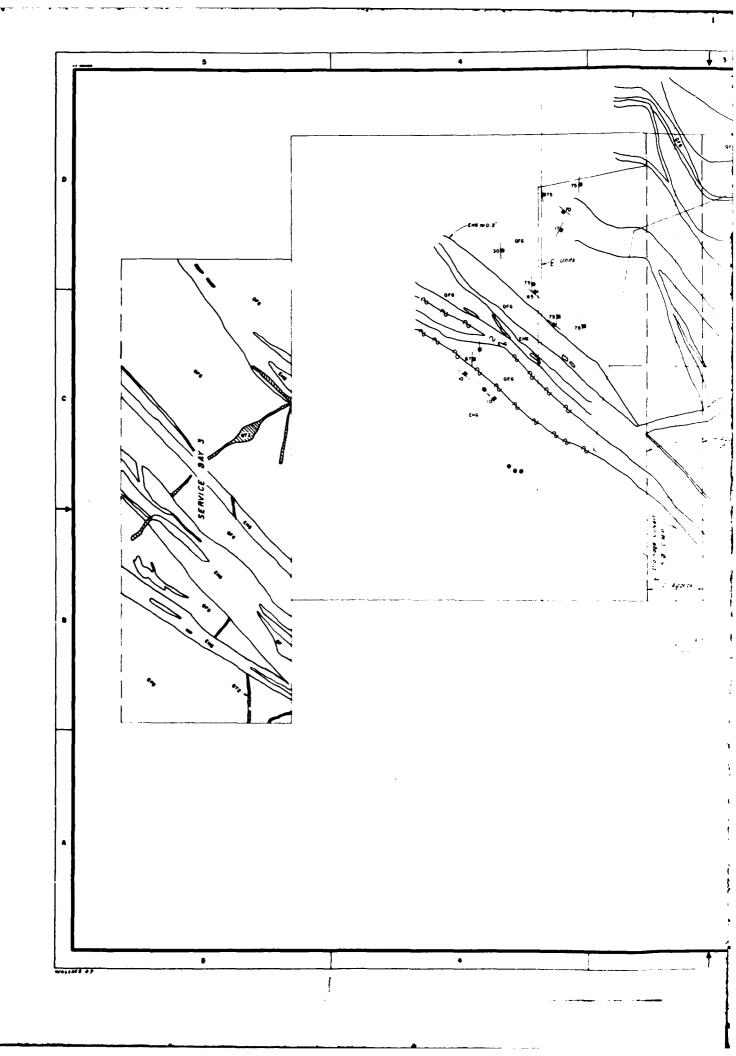


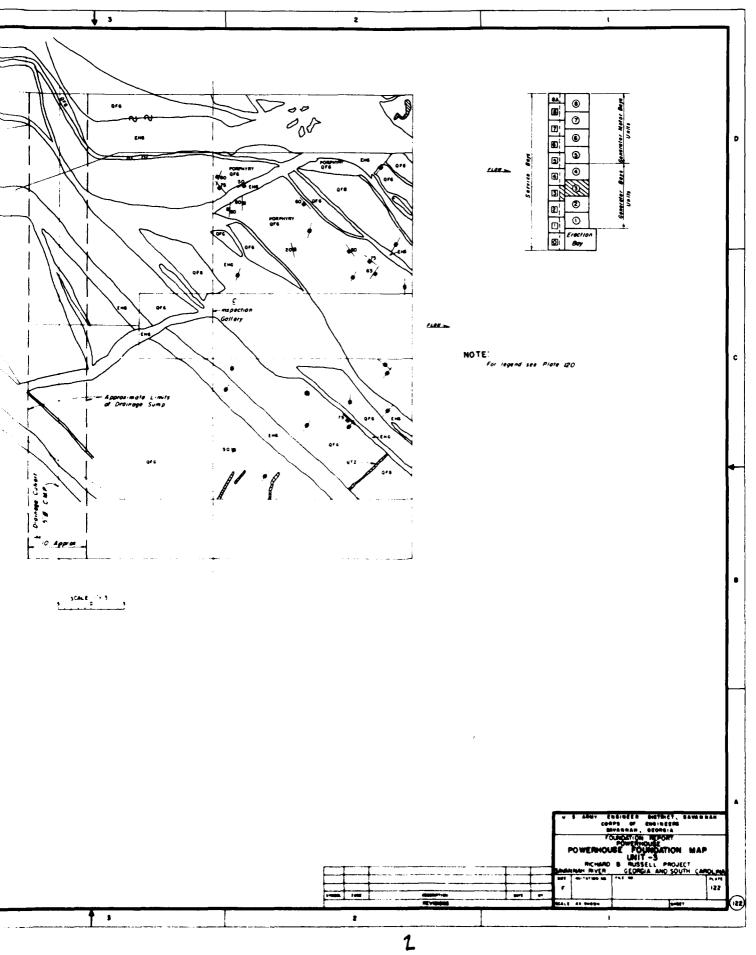


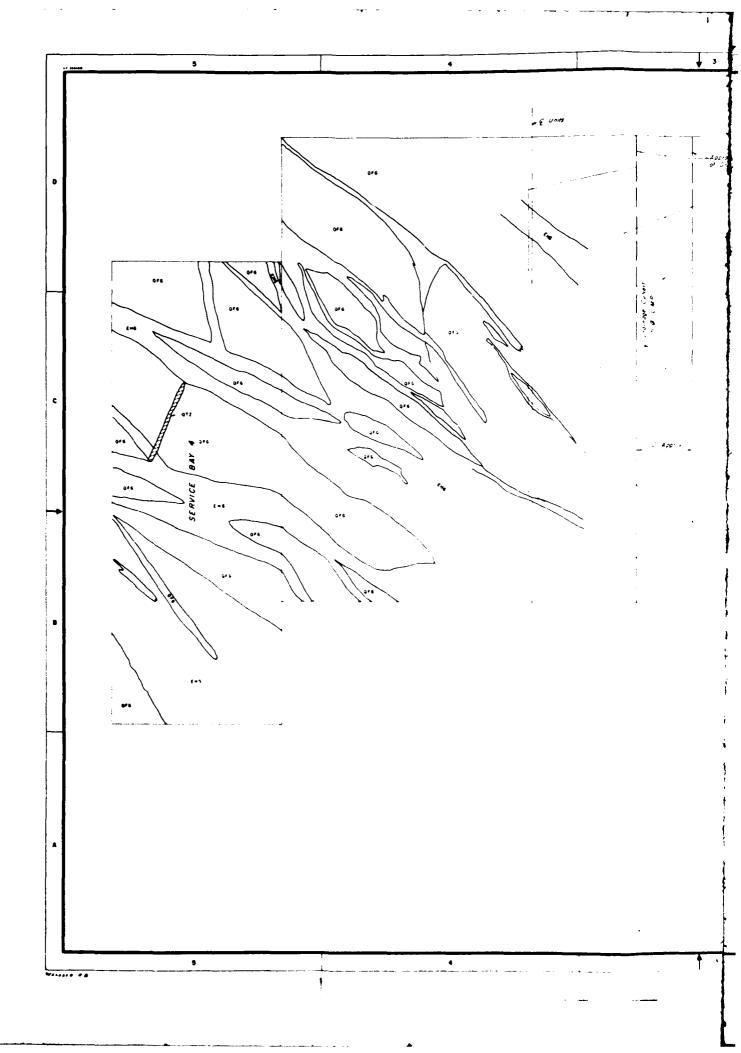


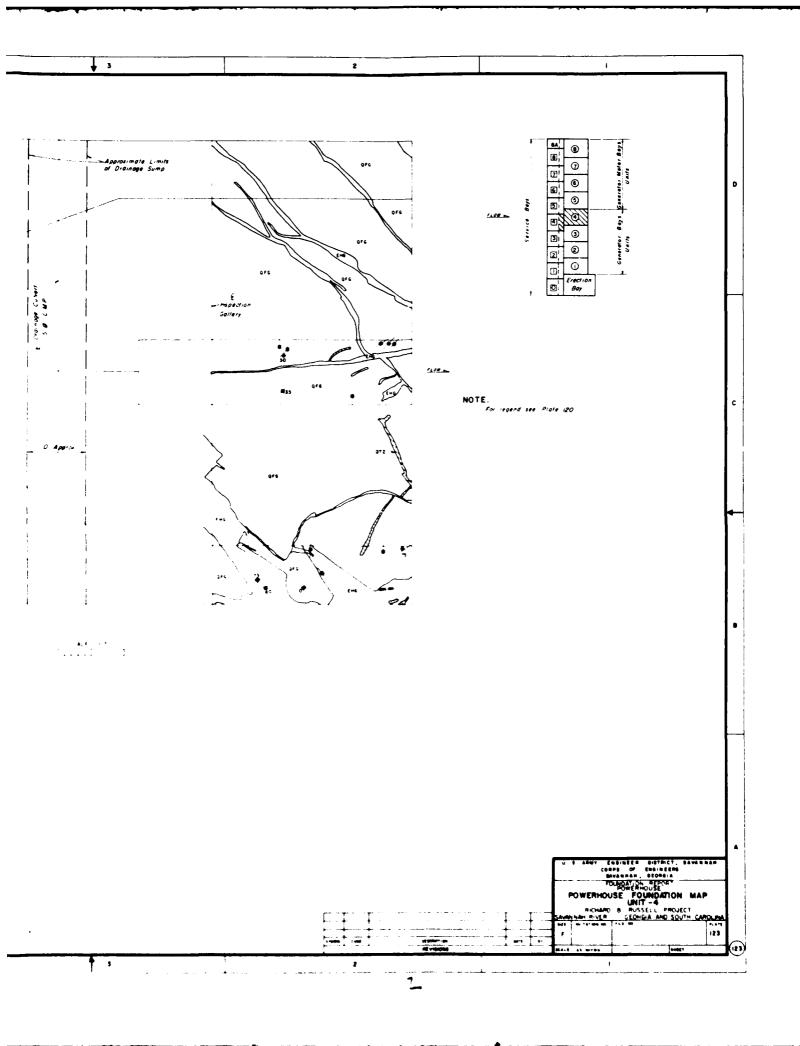


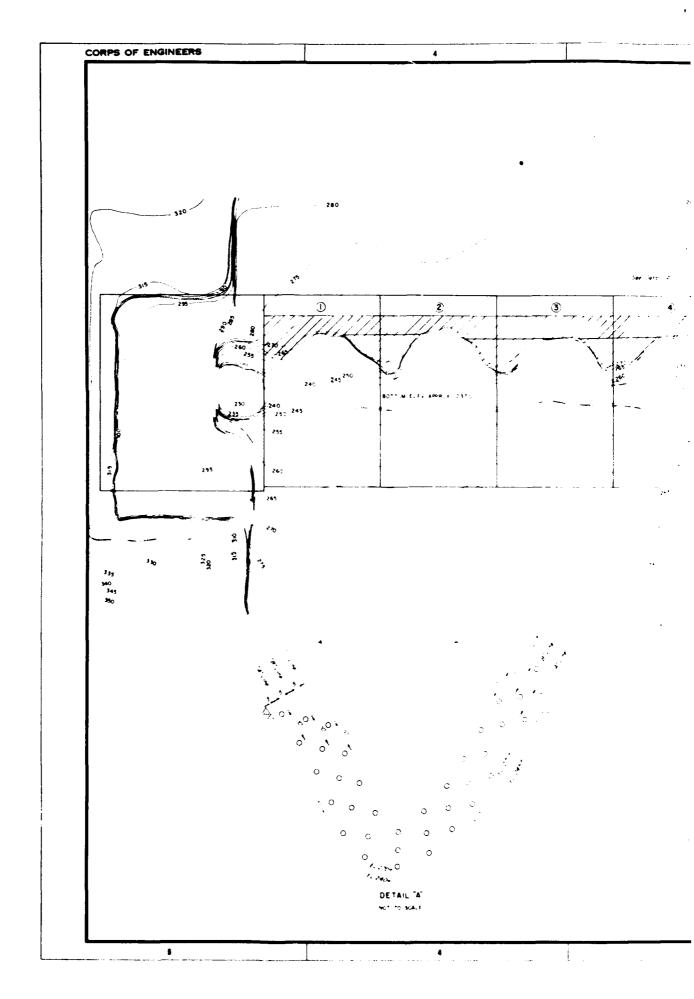


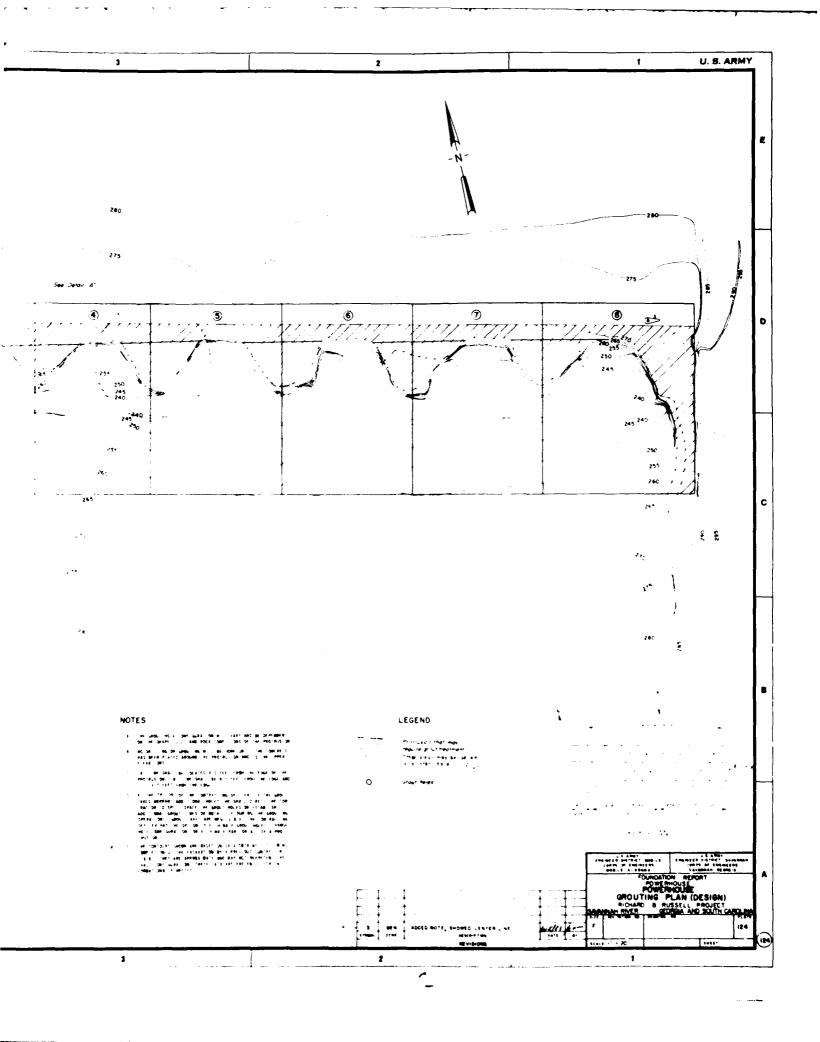


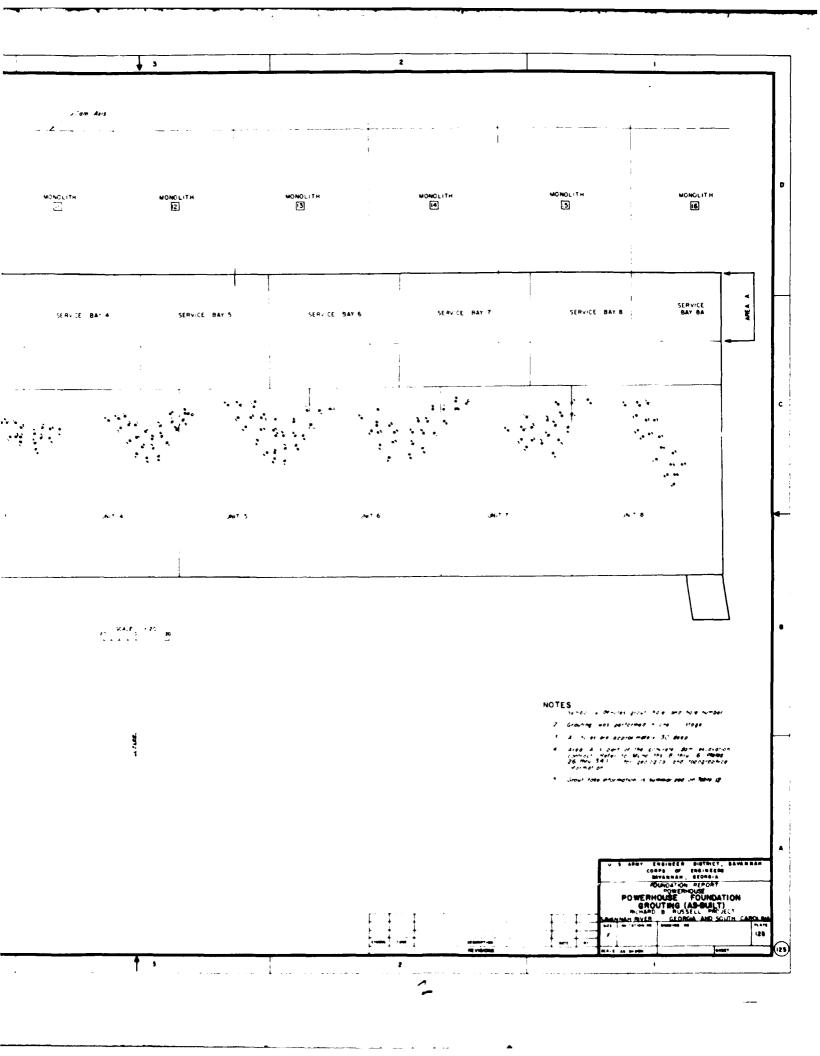




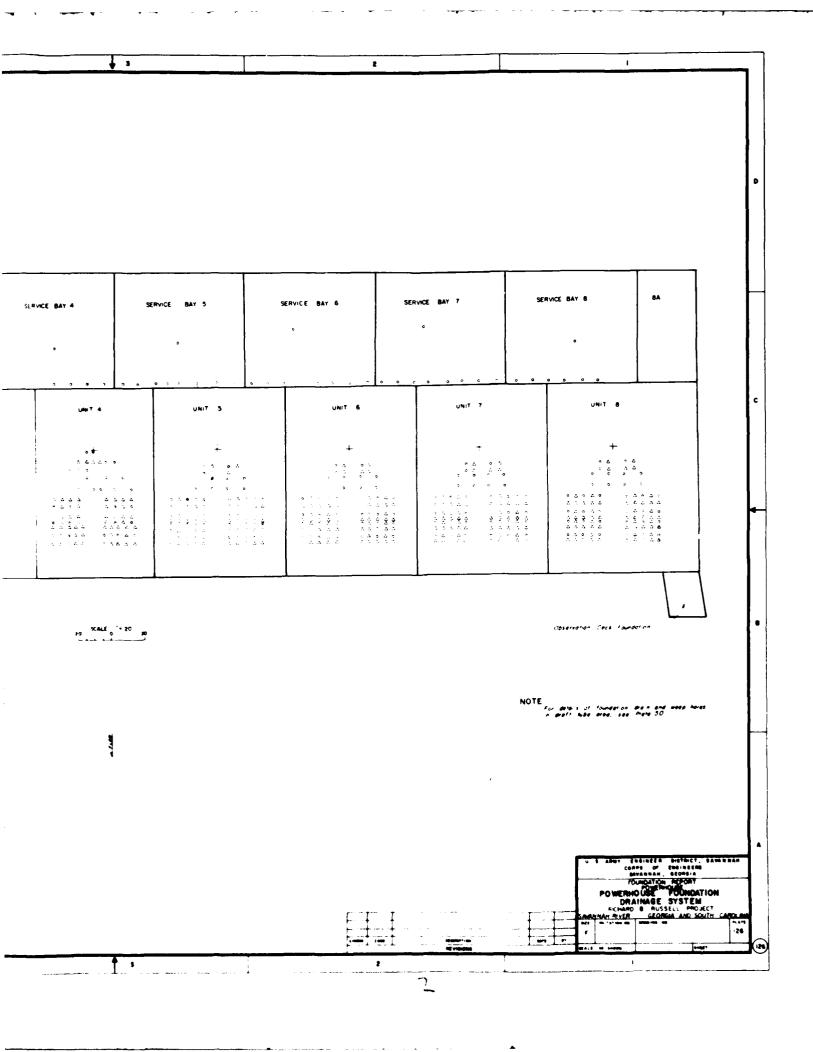








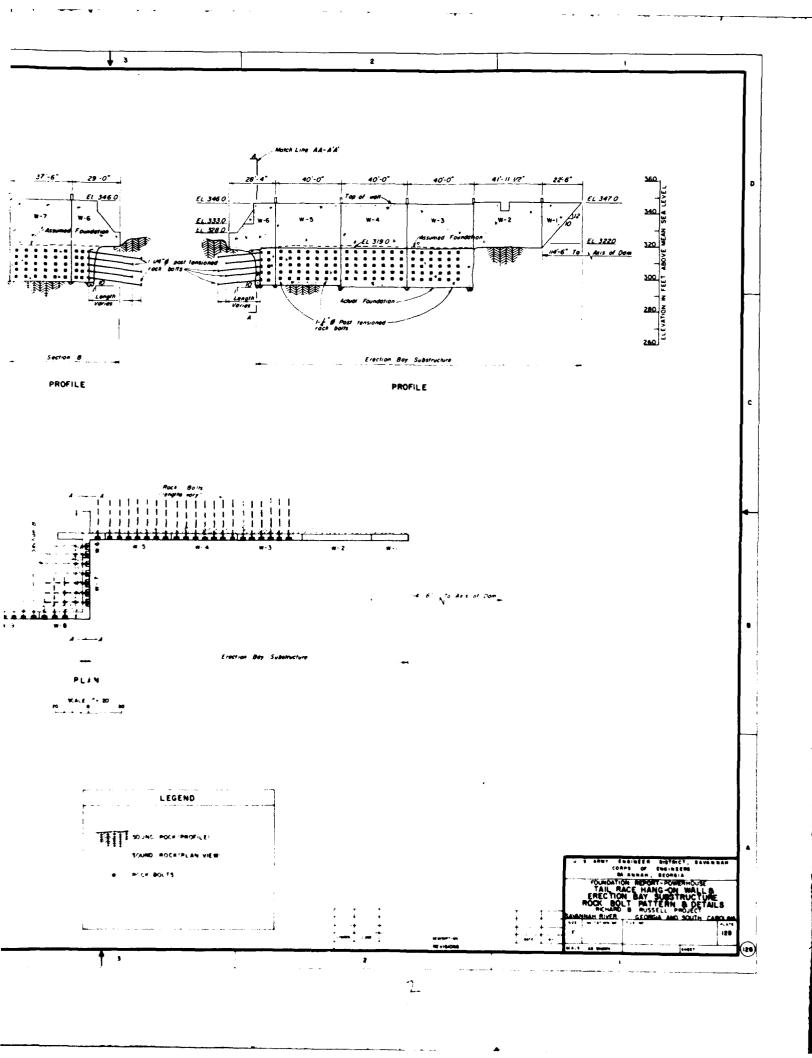
5 SERVICE BAY 4 SERVICE BAY 2 SERVICE BAY 3 SERVICE BAY I SERVICE BAY O UNIT 3 UNIT 2 ERECTION BAY 1 1 4 5 6 1 2 4 6 1 4 5 6 2 4 6 6 2 4 6 6 2 4 6 7 2 6 7 7 . . LEGEND O FOUNDATION CRAIN A meen more + CENTER WOUNT

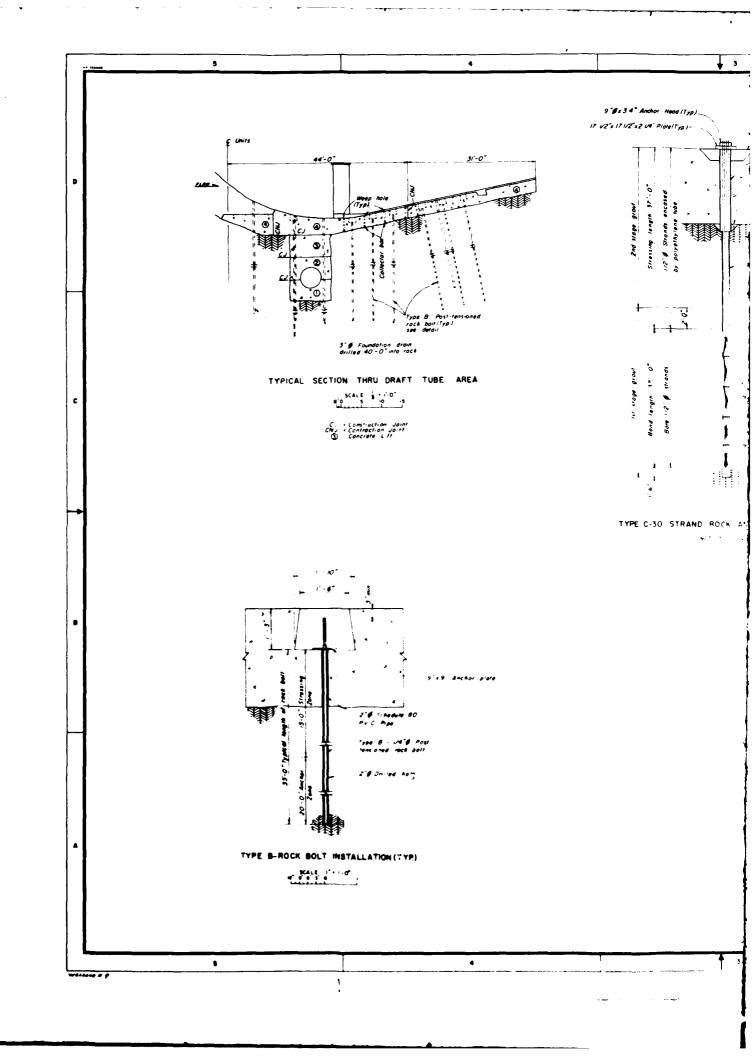


5 4 SERVICE BAY 5 SERVICE BAY 4 SERVICE BAY 2 SERVICE BAY 1 • • 2 ; c UNIT 2 UNIT LEGEND 5 4

3 2 SERVICE BAY 7 SERVICE BAY 6 SERVICE BAY 4 SERVICE BAY 5 ONIT'S 1 2 9 9 5 5 5 5 7 6 7 6 6 3 5 5 **0** 5 7 , , , 3 4 6 6 Jose veren Deck 95.ALE (* 20) U S AMOV ENGINEES SHITHLET, SAME UP AN SURE COMPS OF ENGINEERS SHOWN AND SOUTH COMPS OF THE PROJECT SOUTH OF THE SECONDARY OF 127 . ı

EL 335 0 £L 30001 560 260 PROFILE PROFILE 111 TYPE A ROCK BOLT INSTALLATION (TYPICAL)





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